Lanthanide-doped nanoparticles, their synthesis and uses thereof

Field of the invention

The present invention relates to lanthanide-doped nanoparticles, their synthesis and use thereof. More specifically, the invention relates to lanthanide-doped nanoparticles having superior optical properties, including reduced quenching, effective excitation and a broad bandwidth.

Background of the invention

Lanthanide ions are widely used as the active luminescent material in optical amplifiers and lasers. The major advantages of the lanthanide ions are the long luminescence lifetime and the high quantum efficiency that can be achieved at the major telecommunication wavelengths of 1300 and 1530 nm. The ions are extremely stable under laser action because the optical transitions involve electronic transitions within the ion itself. No chemical bonds are involved so degradation does not occur. The long luminescence lifetime of the lanthanide ions makes laser action relatively easy to achieve with cheap pump sources.

The luminescence of the ions is widely utilized when integrated into inorganic materials, like crystals or glasses. These materials are often fabricated at high temperatures using high-cost materials. For optical glass fiber technology the high costs of integration of the optical components can even exceed the fabrication costs of the components itself. This has disadvantages especially in short range

telecommunication where the optical signals have to be processed in a number of optical components.

To be efficient, optical amplifiers and lasers require complex optical signals that cover a broad wavelength range. In order to achieve this, a large number of optical components are needed. For example, the Er³⁺-doped fiber amplifier (EDFA) is the most used amplifier, because the optical transition of the Er³⁺ ion at 1530 nm coincides with the minimum loss window of optical glass fibers. However, there is a low-loss window. To cover the low-loss window, other lanthanide ions also attract a lot of attention, like Pr³⁺ and Nd³⁺ for amplification around 1300 nm and Tm³⁺ for amplification around 1450 nm. Such complexity of optical componentry causes glass fibre technology to be expensive.

The prior art has shown that planar waveguide structures have significant advantages in this technology. They could for instance profit from current lithography and thus be potentially low cost. In addition, planar structure are potentially easy to integrate with other optical components. A disadvantage in the (polymer-based) integrated wavelength structure is that they have relatively high optical losses. This disadvantage could in principle be overcome by integration of an optical amplifiers to compensate for the optical losses occurring during the manipulation of the data signals.

Polymers are materials that attract a lot of attention for the use in planar waveguides, because of the low cost processing and flexibility these polymers offer in the processing. The prior art L.H. Slooff, A. van Blaaderen, A. Polman, G.A.

Hebbink, S.I. Klink, F.C.J.M. van Veggel, D.N. Reinhoudt; Rare-earth Doped Polymers for Planar Optical Amplifiers J. Appl. Phys. 2002, 91, 3955-3980 incorporated herein by reference) has shown the utility of lanthanide doped organic polymers. Polymers have the advantage that they can be processed at low temperatures.

A further advantage of polymers is that using standard techniques it is relatively easy to make different structures. This makes it possible to integrate a wide variety of optical components like splitters, couplers, multiplexers, and amplifiers. Hence polymers can lower the cost of optical systems. The prior art has shown that organic polymers have a clear potential as low-cost devices, but (intrinsic and extrinsic) optical losses are still relatively high.

Most recently, doped semiconductor nanoparticles have been developed. For example, US Patent Publication No. 20030030067 discloses some Mn.sup.2+ doped semiconductor nanoparticles. In the application the doped semiconductor nanoparticles are contemplated for us in upconversion luminescence ("UCL") materials and methods of making and using same. In the application, any dopant capable of increasing the fluorescence intensity or quantum efficiency of the bulk material or nanoparticle is contemplated for use. With respect to semiconductor nanoparticles, such as ZnS, a dopant capable of increasing fluorescence intensity or quantum efficiency due to the increase of the oscillator strength and the efficient energy transfer from the host to the dopant upon photoexcitation is contemplated. Typically, the dopant will have a high d-d transition rate--e.g. Mn.sup.2+ has a d-d transition of .sup.4T.sub.1.fwdarw.sup.6A.sub.1. Any dopant having a high d-d

transition rate and that is also capable of increasing the luminescence intensity or quantum efficiency of the UCL material is contemplated for use. Hence, the dopants could be broadly classified as rare earth ions--e.g. Tb.sup.3+, Ce.sup.3+ or Eu.sup.3+. The success of these doping procedures has been under discussion. In most of the papers direct prove of the energy transfer of the semi-conductor host to the lanthanide ion is not proven, because an excitation spectrum of the lanthanide ion is missing. O.E. Raola, G.F. Strouse Nano Lett. 2002, 2, 1443., incorporated herein by reference, reported the doping of Eu³⁺ ions in CdSe nanoparticles, starting with Eu²⁺, but no luminescence studies were performed to see if energy transfer from the nanoparticle host to the lanthanide ion takes place. Difficulties in doping most likely arise from the large difference in size between the host cation and the lanthanide ion, the charge mismatch between the cations, and the low affinity of lanthanide ions towards sulfur and selenium.

Problems arise when lanthanide ions are used as the active material in polymer-based optical amplifiers. The luminescence of lanthanide ions, especially the ions emitting at the wavelengths that are of interest for telecommunication, is quenched significantly due to the presence of the organic bonds of the polymer. To increase the luminescence of the lanthanide ions the ion has to be shielded from the polymer environment and one way to do this is by the synthesis of organic complexes, in which an organic ligand is coordinated to the lanthanide ion. Reducing the amount of organic bonds in close proximity to the lanthanide ion has improved the luminescence a little bit, but still quenching is a major constraint to the approach.

It is an object of the present invention to overcome the deficiencies of the prior art.

Summary of the Invention:

The present invention provides photonic materials for use in telecommunications, diagnostics, imaging and lighting, in addition to methods of production of the photonic materials.

A photonic material of the present invention is comprised of a core and an envelope. The core comprises at least one nanoparticle and the nanoparticle comprises an inorganic material and at least one luminescent ion. The inorganic material is suitably selected to bind the luminescent ion. The envelope comprises a suitably selected organic stabilizing layer overlying the core.

In one aspect of the invention, organic stabilizing layer is suitably selected for reducing quenching.

In another aspect of the invention, the organic stabilizing layer comprises a functional group-presenting material and at least one functional group.

In yet another aspect of the invention, the functional group binds an at least one luminescent ion.

In yet a further aspect of the invention, the luminescent ion is a lanthanide ion.

In another aspect of the invention, the functional group-presenting material comprises at least one ligand.

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In yet another aspect of the invention, the ligand is selected to be suitable for ligand exchange reactions.

In another aspect of the invention, the functional group-presenting material is a polymer.

In another aspect of the invention the organic stabilizing layer is self-assembled.

In another aspect of the invention the organic stabilizing layer further comprises one of hole conductors, electron conductors or a suitably selected combination of hole conductors and electron conductors.

In another aspect of the invention the hole conductors comprise heterocyclic compounds.

In another aspect of the invention the heterocyclic compounds are aromatic amines.

In another aspect of the invention the electron conductors are aromatic or heterocyclic compounds with suitably selected reduction potentials.

In another aspect of the invention the aromatic or heterocyclic compounds are one of oxadiazoles, 1,2,4-triazoles, 1,3,5-triazines, quinoxalines, oligo- and polythiophenes and oligo- and polypyrroles.

In another aspect of the invention the envelope is suitably selected for preparing a solgel derived material.

In another aspect of the invention the envelope is suitably selected for solubility in an aqueous environment.

In another aspect of the invention the envelope is suitably selected to permit modification of the functional groups.

In another aspect of the invention the at least one luminescent ion is a lanthanide ion.

In another aspect of the invention the lanthanide ion is selected from the group consisting of Ce, Pr, Nd, Sm, Eu, Gd, Tb, Dy, Ho, Er, Tm and Yb.

In another aspect of the invention the lanthanide ion is selected from the group consisting of Ce, Pr, Nd, Sm, Eu, Gd, Tb, Dy, Ho, Er, Tm, Yb and combinations thereof such that the photonic material emits in the UV-Vis wavelength range.

In another aspect of the invention the lanthanide ions are selected from the group consisting of Ce, Pr, Nd, Sm, Eu, Gd, Tb, Dy, Ho, Er, Tm, Yb and combinations thereof, such that the photonic material emits in the (near-)infrared wavelength range.

In another aspect of the invention the inorganic material is a semiconductor or an insulator, suitably selected to promote sensitized emissions.

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In another aspect of the invention the inorganic material is a semiconductor.

In another aspect of the invention the semiconductor is selected from the group consisting of Al_2S_3 , Ga_2O_3 , In_2O_3 , InP, GaAs, InAs, Ga_2X_3 (X = S, Se, Te), In_2X_3 (X = S, Se, Se,

In another aspect of the invention the semiconductor is Al₂S₃ and the luminescent ion is Eu.

In another aspect of the invention the inorganic material is an insulator.

In another aspect of the invention the insulator is selected from the group consisting of LaPO₄, GdPO₄, YbPO₄, LuPO₄ and LaF₃, LaVO₄, YVO₄. LnPO₄ (Ln = lanthanide or Y), LnVO₄ (Ln = lanthanide or Y), LnX₃ (Ln = lanthanide and X = Cl, Br, I).

In another aspect of the invention the insulator is LaF₃.

In another embodiment of the invention, a suitably selected inorganic shell is provided. The inorganic shell is located between the core and the envelop.

In another aspect of the invention the shell comprises a suitably selected semiconductor.

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In another aspect of the invention the shell comprises a semi-conductor selected from the group consisting of Al_2S_3 , Ga_2O_3 , In_2O_3 , InP, GaAs, InAs, Ga_2X_3 (X = S, Se, Te), In_2X_3 (X = S), X = S, X = S

In another aspect of the invention the shell comprises an insulator.

In another aspect of the invention the insulator comprises LnX₃, wherein X is a halide.

In another aspect of the invention the insulator is LaPO₄.

In another aspect of the invention the insulator is LaF_3 .

In another aspect of the invention the insulator is selected from the group consisting of GdPO₄, LuPO₄, and YPO₄.

In another aspect of the invention the shell further comprises at least one luminescent ion.

In another aspect of the invention the at least one luminescent ion is a lanthanide ion.

In another aspect of the invention the lanthanide ion is selected from the group consisting of Ce, Pr, Nd, Sm, Eu, Gd, Tb, Dy, Ho, Er, Tm and Yb.

In another aspect of the invention the lanthanide ion is selected from the group consisting of Ce, Pr, Nd, Sm, Eu, Gd, Tb, Dy, Ho, Er, Tm, Yb and combinations thereof such that the photonic material emits in the UV-Vis wavelength range.

In another aspect of the invention herein the lanthanide ions are selected from the group consisting of of Ce, Pr, Nd, Sm, Eu, Gd, Tb, Dy, Ho, Er, Tm, Yb and combinations thereof, such that the photonic material emits in the (near-)infrared wavelength range.

In another embodiment of the invention, a method of preparing a photonic material is provided comprising selecting a suitable inorganic material and a suitably luminescent ion, preparing an at least one nanoparticle therefrom to provide a core, and enveloping the core with a suitably selected organic material to provide an envelope.

In another aspect of the invention the method further comprises selecting the organic stabilizing layer to reduce quenching.

In another aspect of the invention, the organic stabilizing layer comprises a functional group-presenting material and at least one functional group and the functional groups are exchanged.

In another aspect of the invention the method further comprises selecting a suitable inorganic material to provide a shell, covering the core with the shell and enveloping the shell with the envelope.

Brief Description of the Figures:

Figure 1: TEM picture of TiO₂:Eu nanoparticles in accordance with an embodiment of the invention, obtained by heating the reaction mixture for 30 min. The inset shows a magnification of a single nanoparticle with lattice fringes.

- Figure 2: (left) Emission spectrum of TiO₂:Eu nanoparticles of Figure 1, in dichloromethane excited at 300 nm. (right) Absorption (dotted line) and excitation (straight line, λ_{em.} 610 nm) spectrum of TiO₂:Eu nanoparticles in dichloromethane.
- Figure 3: Luminescence decay of TiO_2 :Eu nanoparticles of Figure 1 in dichloromethane ($\lambda_{ex.}$ 300 nm).
- Figure 4: FT-IR spectrum of TiO₂:Eu nanoparticles after drying over P₂O₅
- Figure 5: Average luminescence lifetime of Eu³⁺ in TiO₂ nanoparticles after different reaction times.
- Figure 6: Emission ($\lambda_{ex.}$ 300 nm) and excitation ($\lambda_{em.}$ 1060 Nd³⁺, 1530 Er³⁺, and 980 Yb³⁺) spectra of TiO₂ nanoparticles doped with Yb³⁺, Nd³⁺, and Er³⁺. The offset is for clarity.

Figure 7: LED with TiO₂:Eu nanoparticles. The thickness of the layers are not scaled for clarity.

- Figure 8: Current voltage characteristics of the TiO₂:Eu/PMMA LED.
- Figure 9. TEM pictures of (A) LaF₃:Eu, (B) LaVO₄:Eu, and (C) LaPO₄:Eu nanoparticles with the corresponding size distribution. The insets show high magnifications of single nanoparticles in which the crystallinity of the nanoparticles can be seen.

Figure 10. Emission (up) and excitation (down) spectra of LaF₃:Eu, LaVO₄:Eu, and LaPO₄:Eu. An enlargement of the $^5D_0 \rightarrow ^7F_0$ emission peak is shown on the top right.

Figure 11. Normalized emission, λ_{ex} 250 nm (top) and Excitation (bottom) spectra of LnPO₄:Eu particles in methanol. The excitation spectra ($\lambda_{em.}$ 612 nm) are normalized on the 393 nm excitation peak of Eu³⁺.

Figure 12. Luminescence decays (top) of LaF₃:Eu ($\lambda_{ex.}$ 397) in dichloromethane and LaPO₄:Eu ($\lambda_{ex.}$ 260 nm) in methanol, collecting the emission at 591 nm. The fits were obtained by using the model described in the text. The bottom picture shows the distribution of the lifetimes over the particle axis.

Figure 13. Time-resolved emission spectra of LaPO₄:Eu in methanol with a delay of 0.2 ms and 29 ms after the excitation pulse.

Figure 14. Schematic representation of a particle and the representation of the symbols used in the text.

Figure 15. Emission spectra of LaF₃:Ce,Tb particles in CH₂Cl₂ and quinine bisulfate dissolved in 1M H₂SO₄ (λ_{ex.} 282 nm).

Figure 16. Emission (left) and excitation (right) spectra of (A) LaF₃:Nd (dotted line), LaPO₄:Nd (solid line) (B)LaF₃:Er (dotted line), LaPO₄:Er (solid line) (C) LaPO₄:Pr and (D) LaF₃:Ho. The LaF₃ particles were dissolved in CH₂Cl₂ and the LaPO₄ particles in methanol-d4. The emission spectra of the Nd³⁺ doped particles were collected after excitation at 514 nm and the excitation spectra by collecting the emission at 1060 nm. The Er³⁺ doped particles were excited at 488 nm and the emission measured at 1530 nm. For Pr³⁺ and Ho³⁺ the excitation was at 476 and 448 nm, respectively, and the emission at 730 nm and 966 nm respectively. Emission (top) and excitation (bottom) spectra of LaPO₄:Yb particles in methanol.

Figure 17. Emission λ_{ex} . 250 nm (top) and λ_{ex} . 378 nm (middle) spectra of LaPO₄:Yb, Er particles in methanol. The bottom picture shows the excitation

spectrum of the same sample collecting the emission at 1530 nm.

Figure 18. Emission λ_{ex} . 250 nm (top) and λ_{ex} . 378 nm (middle) spectra of LaPO₄:Yb, Er particles in methanol. The bottom picture shows the excitation spectrum of the same sample collecting the emission at 1530 nm.

Figure 19. Emission spectra (top) and excitation spectra (bottom) of LaF₃:Yb,Er with 5% Er³⁺ and different amounts of Yb³⁺. The particles were dissolved in CH₂Cl₂.

- Figure 20A& B: Optical gain of a LaF₃:Nd doped polymer waveguide as a function of pump-power for a 10x10 µm channel pumped at 795 nm.
- Figure 21 Ligands used in the synthesis and for the exchange of LaF₃ nanoparticles
- Figure 22: ¹H NMR spectra of LaF₃ nanoparticles synthesized with ligand 2 (left) and after exchange with ligand 3 (right).
- Figure 23: ¹H NMR spectrum of LaF₃ nanoparticles synthesized with ligand 1 before (left) and after (right) the exchange with oleic acid. The left picture shows an enlargement of the broad peak of the protons next to the dithiophosphate group.
- Figure 24: Synthesis route for functional dithio ligands: i: K₂CO₃, acetonitril, reflux, yield: 99%; ii: P₂S₅, NH₃, yield: 60%.
- Figure 25: HNMR of LaPO₄, surface modified by a reaction with dodecanol.
- Figure 26: Emission spectra of LaPO₄:Eu nanoparticles in methanol with tetramethylammonium hydroxide (TMAH) and after the surface reaction with dodecanol. The difference in intensity is for clarity.

Figure 27: Schematic of a core-shell nanoparticle, with a LaCeF₃:Tb core and a LaF₃ shell.

- Figure 28: TEM pictures of LaCeF₃Tb nanoparticles with the corresponding size distribution histograms, core (top) and core-shell (down).
- Figure 29. Luminescence decay curves of the ⁵D₀ (left) and ⁵D₁ (right) level of Eu³⁺ in LaF₃ nanoparticles with a different amount of shell material in toluene.
- Figure 30: Emission of the 5D_2 level of Eu $^{3+}$ in core-shell nanoparticles. The spectra are normalized to the emission of the $^5D_0 \rightarrow ^7F_1$ emission peak. The offset is for clarity.
- Figure 31: 5 μm wide PMMA/LaF₃:Nd ridge structures on a SiO₂ wafer.
- Figure 32: Emission spectrum of LaF₃:Nd nanoparticles in solution and in the PMMA waveguide after excitation at 578 nm.
- Figure 33: Setup for the measurement of optical amplification on the waveguides.
- Figure 34: Composition of a SU-8 negative photoresist solution.

Figure 35: Waveguides of SU-8 doped with 20% of LaF₃:Nd, (left) top few of waveguide channels, (right) cross-sectional view of a diced 5x10 µm channel.

- Figure 36: Chip losses of the pigtailed SU-8/LaF₃:Nd waveguide.
- Figure 37: Nd³⁺ emission spectrum measured at the end of the SU-8/LaF₃:Nd waveguide after pumping at 795 nm.
- Figure 38: Amplification calculated for (left) the PMMA 10x3.3 μm waveguide for a pump power of 12 mW at 578 nm and the 10x10 μm SU-8 waveguide for a pump power of 5 mW at 795 nm.
- Figure 39: Optical gain of a LaF3:Nd doped polymer waveguide as a function of pump-power for a $10x10~\mu m$ channel pumped at 795~nm.
- Figure 40: Optical gain of a LaF₃:Nd polymer waveguide as a function of the dimensions of the waveguide. The pump-power is fixed at 100 mW.
- Figure 41. Optical gain calculated for different luminescence lifetimes. The pump power is 100 mW and the channel dimensions are 10 by 10 mm.

Description of the Invention:

Definitions

The term "nanoparticles" as used herein, can also refer to nanoclusters, clusters, particles, dots, quantum dots, small particles, and nanostructured materials.

When the term "nanoparticle" is used, one of ordinary skill in the art will appreciate that this term encompasses all materials with small size and often associated with quantum size effects, generally the size is less than 100 nm. As a new type of materials, nanoparticles are luminescent with high quantum efficiency and are good candidates for upconverters, ELDs and optical amplifiers. It is known that oscillator strength is a very important optical parameter that determines the absorption crosssection, recombination rate, luminescence efficiency, and the radiactive lifetime in materials. The oscillator strength of the free exciton is described by 1 f e x = 2 m E | | 2 | U (0) | 2 where m is the electron mass, .DELTA.E is the transition energy, .mu. is the transition dipole moment, and .vertline.U(0).vertline..sup.2 represents the probability of finding the electron and hole at the same site (the overlap factor). In nanostructured materials, the electron-hole overlap factor increases largely due to the quantum size confinement, thus yielding an increase in the oscillator strength. The oscillator strength is also related to the electron-hole exchange interaction that plays a key role in determining the exciton recombination rate. In bulk semiconductors, due to the extreme dislocation of the electron or hole, the electron-hole exchange

interaction term is very small; while in molecular-size nanoparticles, due to the confinement, the exchange term chould be very large.

General Methods. The spectra of Eu^{3+} were measured at a resolution of 1.8 nm and the spectra in the near-infrared were measured at a resolution of 16.2 nm for Er^{3+} and 8.1 nm for Nd^{3+} and Yb^{3+} . The quantum yield of the TiO2:Eu nanoparticles was determined by comparing the emission intensity with the emission intensity of a solution of ruthenium tris bipyridine with the same optical density at the excitation wavelength (λ_{ex} 310 nm). The quantum yield of ruthenium tris bipyridine in deoxygenated water is 4.2 %. For the electroluminescence measurements the same instrument as for the photo-physical measurements was used.

Nanoparticle synthesis. Nanoparticles of lanthanide-doped TiO2 were prepared by adopting a literature procedure from Jones, A. C.; Leedham, T. J.; Wright, P. J.; Crosbie, M. J.; Fleeting, K. A.; Otway, D. J.; O'Brien, P.; Pemble, M. E. J. Mater. Chem. 1998, 8, 1773, all incorporated herein by reference. TiO₂ particles were thermal decompositon of Ti(OPrⁱ)₃(dmae) formed the (dmae dimethylaminoethoxide) in hot trioctylphoshine oxide (TOPO). The titanium precursor was synthesized as follows: lanthanide-doped nanoparticles were prepared by dissolving 11 mg of EuCl₃ · 6H₂O in 5 mL of methanol followed by the addition of 3.5 g of TOPO and evaporation of the methanol and water in vacuo. The resulting solution was heated to 315 °C under a nitrogen atmosphere and 200 µl of the titanium precursor was injected, followed by further heating at 300 °C for 10-240 min. The reaction mixture was cooled to 60 °C and the nanoparticles were precipitated by the addition of methanol. The product was separated by centrifugation and further

purified by washing with methanol. The resulting lanthanide-doped particles are dispersable in apolar solvents like toluene, dichloromethane, and chloroform, to form optically clear solutions.

LED fabrication. LEDs with TiO₂:Eu nanoparticles were prepared on ITO cover glass plates. The ITO layer was first cleaned by sonication in water and ethanol, followed by a treatment in an oxygen plasma for 5 min. On top of the ITO a 80 nm thick layer of PEDOT (Aldrich) was spin-coated at 4000 rpm. The layer was dried over P₂O₅ in vacuum at 80 °C for 2 days. On top of this layer a 1 % solution of TiO2/PMMA 3/1 in chloroform was spincoated at 1500 rpm to form a 100 nm thick layer. This layer was also dried over P₂O₅ in vacuum at 80 °C for 1 day. The devices were then transferred to a glove box with a dry nitrogen atmosphere and on top of the layers a drop of InGa eutectic was placed. On the drop of InGa eutectic a copper contact wire was placed and the device was protected from the atmosphere by a layer of epoxy glue. The glue was cured at 80 °C for 2 hours.

LaF₃ nanoparticle synthesis. All chemicals were used as received without further purification. The lanthanide salts were purchased from Aldrich or Acros in the highest purity available (at least 99.9%). Octadecanol (99%), sodium fluoride (99%), and tris(ethylhexyl)phosphate (98%) were purchased from Fluka, phosphorous pentasulfide (99%) from Aldrich and trioctylamine (98%) from Acros. Oleic acid and octadecanol were purchased from Fluka. Phosphorus oxychloride was purchased from Acros. THF was freshly distilled from Na/benzophenon. ¹H NMR measurements on nanoparticles were performed on undoped nanoparticles to exclude the influence of the magnetic moment of some of the luminescent lanthanide ions.

LaPO₄ particles were synthesized using a literature procedure (Riwotzki, Meyssamy, Kornowski and Haase J. Phys Chem, 104, 2824 incorporated herein by reference). Briefly, 1 mmol of the hydrated LnCl₃ (Ln = La³⁺, Eu³⁺, Nd³⁺, Er³⁺ or Pr³⁺) is first dissolved in 10 ml of methanol followed by the addition of 6 ml tris(ethylhexyl)phosphate and evaporation of the methanol under vacuum. The solution is heated to 70°C under vacuum to remove water. A solution of 1 mmol orthophosphoric acid and 3 mmol trioctylamine in 3 ml of tris(ethylhexyl)phosphate was added and the reaction mixture was heated at 200°C under argon. After 40 hours the solution was cooled to room temperature and methanol was added to precipitate the nanoparticles. The particles were separated by centrifugation and washed two times with methanol. The particles were soluble in methanol by the addition of tetramethylammonium hydroxide. The particles could be precipitated by the addition of ethyl acetate and separated by centrifugation. Particles obtained in this way were surface tetramethylammonium groups and some of the coated with bis(ethylhexyl)phosphate chains.

The ligand ammonium dioctadecyldithiophosphate used in the synthesis of the LaF₃ particles was prepared by heating P₂S₅ (0.02 mol) and octadecanol (0.07 mol) at 75°C for 3 hours. The resulting suspension was cooled to room temperature followed by the addition of dichloromethane and filtration. The solvent was evaporated and the residue taken up in hexane and ammonia was bubbled through this solution. The resulting precipitate was separated by filtration, washed with hexane and dried. ¹H NMR (DMSO d₆): δ 7.2-6.9 (broad, 4H), δ 3.7 (dt, J= 8.1, 6.6 Hz, 4H), δ 1.5-1.4 (m, 4H), δ 1.35-1.1 (m, 60H), δ 0.84 (t, J= 6.2 Hz, 6H); MS (MALDI-TOF) m/z = 634.0

[(M-NH₄⁺)⁻, calcd. for C₃₆H₇₄O₂PS₂: 633.5]; Anal calcd. for C₃₆H₇₈NO₂PS₂: C, 66.31, H, 12.06, N, 2.15, S, 9.83. Found C, 66.50, H, 12.24, N, 2.27, S, 9.56.

The LaF₃ nanoparticles were prepared by heating a solution of 614 mg (0.95 mmol) ammonium di-n-octadecyldithioposhate and 126 mg (1 mmol) NaF in 35 ml ethanol/water at 75°C. A solution of La(NO₃)₃.6H₂O and Eu(NO₃)₃.6H₂O (1.33 mmol total) in 2 ml of water was added dropwise and the solution was stirred at 75°C for 2 hours and then cooled to room temperature. The precipitate was separated by centrifugation and was washed subsequently with water and ethanol. The particles were further purified by dispersing in dichloromethane and precipitation by the addition of ethanol. After separation by centrifugation the particles were dried in vacuum over P₂O₅ for two days. After drying the particles, they are soluble in apolar solvents like chloroform, dichloromethane, and toluene.

TEM images were collected on a Philips CM 30 Twin FTEM, operating at 300 kV. Samples were prepared by evaporating a drop of a diluted particle dispersion, in methanol in the case of LaPO₄ and dichloromethane in the case of LaF₃ particles, on a carbon coated 200 mesh copper grid. ¹H NMR spectra were recorded at 25°C on a Varian Inova 300 NMR spectrometer. Elemental analyses were carried out with a model 1106 Carlo-Erbu Strumentazione elemental analyzer. X-ray fluorescence was carried out on a Philips PW 1489 spectrometer using LaF₃, Li₃PO₄, Eu₂O₃, Nd₂O₃, Er₂O₃, Pr₆O₁₁, and Ho₂O₃ as the standards. Infrared spectra were recorded on a FT-IR Perkin Elmer Spectrum BX spectrometer. Mass spectra were recorded using a Perkin Elmer/Perspective biosystems Voyager-DE-RP MALDI-TOF mass spectrometer. Photoluminescence measurements in the visible were done with an Edinburgh

Instruments FS/FL instrument with a 450 W Xe arc lamp and a red sensitive Peltier element cooled Hamamatsu R955 PMT. The emission spectra were corrected for the detector response and the excitation spectra were corrected for the lamp intensity. The NIR spectrum of Ho³⁺ was measured with the same instrument using a liquid nitrogen cooled Ge detector. The other NIR spectra were measured by exciting the samples with a CW Ar⁺ ion laser operating at 488 nm (Er³⁺), 476 nm (Pr³⁺), and 514 nm (Nd³⁺). The continuous light was modulated with an acousto-optic modulator and focused on the sample in 1x1 cm² quartz cuvettes. The signal was focused with a 20 cm lens on a monochromator and detected at the monochromator exit with a liquid nitrogen cooled Ge detector (Northcoast) or a PMT (AgOCS) for the spectral region between 700-1000 nm. The signal of the detector was fed to a lock-in amplifier. The spectra of the near-infrared emitting ions were measured with a resolution of 6 nm and of the visible emitting ions with a resolution of 0.9 nm.

Luminescence quantum yields were determined by comparing the luminescence intensity of a particle solution with a solution of quinine bisulfate in 1M H_2SO_4 with approximately the same absorption at the excitation wavelength of 274 nm . The quantum yield was calculated from the equation $\phi_{\text{sample}} = \frac{n^2_{\text{sample}} I_{\text{sample}} A_{\text{ref}}}{n^2_{\text{ref}} I_{\text{ref}} A_{\text{sample}}} \phi_{\text{ref}}$ with n the refractive index, A the absorption, and I the emission intensity and taking the quantum yield of quinine bisulfate as 54.4%.

Characterization of LaF₃ nanoparticles. The doping concentration in all particles was 5% of the luminescent ion compared to the total concentration of lanthanide ions, to give a molar ratio of La³⁺/Ln³⁺ of 19/1, unless stated otherwise. Particles of LaPO₄

were synthesized using a literature procedure from the hydrated LnCl₃ salts and phosphoric acid. The growth of the particles is controlled by the coordinating solvent tris(ethylhexyl)phosphate and protonated trioctylamine. During the reaction some of the solvent is hydrolyzed and the resulting bis(ethylhexyl)phoshate binds to the surface as can be seen by ¹H NMR spectroscopy. The LaF₃ nanoparticles were synthesized using ammonium dioctadecyldithiophosphate as ligand. To a solution of this ligand and NaF in a water ethanol mixture was added dropwise a solution of the Ln(NO₃)₃ salts in water. The dithiophosphate head-group of the ligand coordinates weakly to the lanthanide ions allowing the growth of the nanoparticles, but coordinates strongly enough to keep the particles from aggregating.

Oleic acid and octadecanol were purchased from Fluka. Phosphorus oxychloride was purchased from Acros. THF was freshly distilled from Na/benzophenon. ¹H NMR measurements on nanoparticles were performed on undoped nanoparticles to exclude the influence of the magnetic moment of some of the luminescent lanthanide ions.

$(CH_3(OCH_2CH_2)_nO)_2PS_2^-NH_4^+(1)$:

A mixture of P₂S₅ (0.02 mol) and poly(ethyleneglycol)methylether (molecular weight ca. 550 g/mol) (0.07 mol) was heated to 75°C for 3 hours. The suspension was cooled to room temperature, followed by the addition of 50 ml of dichloromethane. The solution was filtered, followed by the removal of solvent by rotary evaporation. The remaining residue was dissolved in 50 mL of hexane, and ammonia gas bubbled through the solution for 5-10 minutes. The solvent was removed by rotary evaporation leaving behind a viscous white/yellow product (mw. 1210.97g/mol).

Synthesis of nanoparticle with ligand 1 (GaF₃:Eu, LaF₃:M, M = Eu³⁺, Er³⁺):

At room temperature, a solution of 1 (0.95 mmol, 1.15g)) in 30 mL of THF was stirred for 15minutes, followed by the addition of NaF (3 mmol) in 7ml of H₂O. The solution was stirred for 15 minutes, followed by the dropwise addition of La(NO₃)₃·6H₂O and M(NO₃)₃·xH₂O (1.26 mmol and 0.04 mmol respectively) in 2 mL of H₂O. The resulting solution was then refluxed at 97°C for 3-12 hours, after which it was cooled to room temperature. The resulting precipitate was separated by centrifuge, washed three times with THF, and then dried under vacuum. :

Ligand 3 (C₁₈H₃₇O)₂PO₂ NH₄⁺. A solution of 5 g (18.5 mmol) octadecanol in 20 ml of anhydrous THF was slowly dropped to 1.42 g (9.2 mmol) of OPCl₃ at -15 °C under Ar. The solution was slowly allowed to warm to room temperature and stirred for two hours. 1 ml of water was added and the solution was vigorously stirred for 1 hour. 100 ml of ethyl acetate was added and the organic phase was washed three times with 1 N HCl. The organic layer was dried over MgSO₄ and evaporated to dryness. 20 ml of hexane was added and NH₃ was bubbled to the solution. The precipitate was filtered, washed with hexane, and dried. To substitute the ammonium counterion for a proton, the product was taken up in ethyl acetate and washed three times with 1 N HCl. The organic layer was dried over MgSO₄ and evaporated to dryness. A white solid remained, yield 2.9 g, 53 %, m.p. 71-73 °C. ¹H NMR (CDCl₃): δ 4.1 (dt, *J* = 7.0, 6.6 Hz, 4H), 1.7 (m, 4H), 1.3 (m, 60 H), 0.9 (t, *J* = 6.6 Hz, 6H); MS (FAB) m/z = 602.5 [(M⁺), calcd. for C₃₆H₇₅O₄P: 603.0]; Anal calcd. for C₃₆H₇₅O₄P: C, 71.71; H, 12.54. Found: C, 71.45; H, 12.80.

Compound 5a p-CH₃OOC-Ph-O(CH₂)₁₁OH was synthesized by refluxing a mixture of 10 g (39.3 mmol) bromoundecanol, 7.78 g (51 mmol) of methyl-4-hydroxy benzoate, and 7.06 g (51 mmol) K_2CO_3 in 100 ml of acetonitril overnight. The mixture was cooled to room temperature and 300 ml of ethylacetate was added. This solution was washed with 1 N NaOH three times and one time with water. The organic layer was dried over MgSO₄, filtered, and the organic solvent removed under vacuum. A white solid remained, that was recrystallized from ethylacetate, yield 99%, m. p. 69-71 °C. ¹H NMR (CDCl₃): δ 8.1 (d, J = 8.8 Hz, 2H), 6.9 (d, J = 8.8 Hz, 2H), 4.0 (t, J = 6.2 Hz, 2H), 3.9 (s, 3H), 3.6 (dt, J = 6.6, 4.4 Hz, 2H), 1.8 (m, 2H), 1.6-1.4(m, 16H); MS (FAB) m/z = 323.2 [(M+H)⁺, calcd. for $C_{19}H_{31}O_4$: 323.2]; Anal calcd. for $C_{19}H_{30}O_4$: C, 70.77; H, 9.38. Found: C, 70.88; H, 9.55.

Ligand 5 p-CH₃OOC-Ph-O(CH₂)₁₁O)₂PS₂ NH₄ was synthesized by heating 9.72 g (30 mmol) of compound 5a with 1.91 g (8.6 mmol) of P₂S₅ at 80 °C for 3 hours. The workup of this reaction was the same as for ligand 2. The product was a white solid obtained in a yield of 73 %, m. p. 81-83 °C. ¹H NMR (acetone- d_6): δ 7.9 (d, J = 9.2, 4H), 7.0 (d, J = 9.2, 4H), 4.1 (t, J = 6.6, 4H), 3.9 (dt, J = 8.4, 6.6, 4H), 3.8 (s, 6H), 1.8 (m, 4H), 1.5-1.3 (m, 32H); MS (FAB) m/z = 737.5 [(M-NH₄), calcd. for C₃₈H₅₈O₈PS₂: 737.0]; Anal calcd. for C₃₈H₆₂O₈PS₂N: C, 60.37; H, 8.27; N, 1.85; S, 8.48. Found: C, 59.87; H, 8.72; N, 1.95; S, 8.37.

Compound 6a: p-Cl-Ph-O(CH₂)₁₁OH was synthesized by refluxing 7.06 g (51 mmol) K_2CO_3 , 6.6 g (51 mmol) of 4-chlorophenol, and 10 g (39.3 mmol) bromoundecanol in 100 ml of acetonitril overnight. The workup of the reaction mixture was the same as

for compound **5a**. The product was a white solid recrystallized from hexane and obtained in a yield of 98%, m. p. 42-43 °C. ¹H NMR (CDCl₃): δ 7.3 (d, J = 9.2, 2H), 6.8 (d, J = 9.2, 2H), 3.9 (t, J = 6.6, 2H), 3.7 (dt, J = 6.6, 5.1, 2H), 1.8 (m, 2H), 1.6-1.3 (m, 16H); MS (FAB) m/z 298.3 [M⁺, calcd. for C₁₇H₂₇O₂Cl: 298.2; Anal calcd. for C₁₇H₂₅O₂Cl: C, 68.79; H, 8.49. Found: C, 68.54; H, 8.35.

Ligand 6 (p-Cl-Ph-O(CH₂)₁₁O)₂PS₂⁻ NH₄⁺ was synthesized by heating 9.7 g (32.5 mmol) of compound 6a and 2.06 g (9.3 mmol) P₂S₅ at 80 °C for 3 hours. The workup of this reaction mixture was the same as for Ligand 5. The product was a white solid obtained in 70 % yield, m. p. 57-60 °C. ¹H NMR (CDCl₃) δ 7.2 (d, J = 9.2, 4H), 6.8 (d, J = 8.8, 4H), 4.0 (dt, J = 8.4, 6.9, 4H), 3.9 (t, J = 6.6, 4H), 1.8-1.7 (m, 8H), 1.4-1.3 (m, 28H); MS (FAB) m/z = 689.2 [(M-NH₄)⁻, calcd. for C₃₄H₅₂O₄PS₂Cl₂: 689.2]; Anal calcd. for C₃₄H₅₆O₄PS₂NCl₂: C, 57.61; H, 7.96; N, 1.98; S, 9.05. Found: C, 57.42; H, 8.05; N, 2.04; S, 9.20.

Exchange reaction on LaF₃ nanoparticles. To a solution of 100 mg of LaF₃ nanoparticles in 5 ml dichloromethane was added 100 mg of ligand 3 or 4 and 100 μ l of triethylamine. This solution was stirred overnight and methanol was added to precipitate the nanoparticles. The precipitate was separated by centrifugation and washed repeatedly with polar solvents like methanol, ethanol, or acetone, until no free ligand was observed in the 1H NMR spectrum.

Surface modification of LaPO₄ nanoparticles. The surface phosphate groups of the nanoparticles were first reacted with phosphorus oxychloride by heating 100 mg of nanoparticles in 5 ml of OPCl₃ for two hours at 120 °C. After two hours the solution

was cooled to room temperature and the OPCl₃ was evaporated under vacuum followed by the addition of 5 ml of toluene containing 1 ml of dodecanol. This solution was refluxed for two hours and then cooled to room temperature. To remove excess dodecanol the nanoparticles were precipitated by the addition of 20 ml of methanol and separated by centrifugation. After washing a few times with methanol the nanoparticles were soluble in apolar solvent like toluene, dichloromethane, and chloroform.

Core-shell nanoparticles. LaF₃ core-shell nanoparticles were synthesized similar to the core nanoparticles described above. NaF was substituted by NH₄F because of an increased solubility in water. The core of the nanoparticles was synthesized by the dropwise addition of the lanthanide salts to the NH₄F and ligand 1 in 35 ml of a water/ethanol mixture at 75 °C. This mixture was stirred for 10 min followed by the shell growth. To determine the best way of growing a shell around a core, several methods were tested.

method 1. After the synthesis of the core nanoparticles 3 mmol of NH₄F was added drop-wise followed by the drop-wise addition of 1.33 mmol of La(NO₃)₃.

method 2. The synthesis of the core nanoparticles was started with 6 mmol of NH₄F and to this solution was first added the lanthanide salts for the core, followed after 10 min by the addition of La(NO₃)₃ for the shell.

method 3. After the synthesis of the core, 3 mmol of NH₄F and 1.33 mmol of La(NO₃)₃ in water were added alternately in small portions.

After the addition of the shell material the solution was stirred for 2 more hours at 75 °C, after which it was cooled down to room temperature. The workup of the nanoparticle product was the same as described above.

Spincoating of the polymer solutions was performed under standard cleanroom conditions. The thicknesses of the polymer layers were measured using a Dektak surface profiler. Arch, 907/17 was used as the positive photoresist and spincoated at a speed of 4000 rpm for 20 s and a prebake temperature of 95 °C. The photoresist was developed in standard OPD developer for 50 s. Reactive ion etching was performed using a Elektrotech PF340 etcher. The endfaces of the SU-8 waveguide were made by dicing the substrate using a Disco DAD-321 Dicing Saw.

LaF₃:Nd nanoparticles. LaF₃:Nd nanoparticles were synthesized as described above. PMMA layers were spincoated from chlorobenzene and ligand 2 was used to make LaF₃:Nd nanoparticles that are soluble in chlorobenzene. SU-8 is a commercial product with γ -butyrolactone as the solvent. Nanoparticles synthesized with ligand 2 are soluble in γ -butyrolactone, but for mixing with the SU-8 solution, a mixture of ligand 2 and 5 in the ratio 1:2 was found to give the best solubility properties. The nanoparticle-doped solution was made by dissolving the nanoparticles in γ -butyrolactone and mixing this solution with commercially available SU-8 25 (solid content 63%), to end up with a solid content of 52%.

Exchange reaction on LaF₃ nanoparticles. To a solution of 100 mg of LaF₃ nanoparticles in 5 ml dichloromethane was added 100 mg of ligand 3 or 4 and 100 μ l of triethylamine. This solution was stirred overnight and methanol was added to precipitate the nanoparticles. The precipitate was separated by centrifugation and washed repeatedly with polar solvents like methanol, ethanol, or acetone, until no free ligand was observed in the ¹H NMR spectrum.

Preparation of SiO₂ thin films incorporated with LaF₃:Er nanoparticles involve two steps which include preparation of water soluble LaF₃:Er-citrate nanoparticles and then reacting the aqueous solution of these particles with tetraethylorthosilicate (TEOS),ethanol mixture in presence of hydrochloric acid.

Preparation of water soluble LaF3:Er-citrate nanoparticles

An aqueous solution made by dissolving 0.126g of NaF and 2.0 g of citric acid, in 35 ml of water was neutralized with NH₄OH till the pH reaches 7. This solution was then heated to 75°C. At this temperature, 2 ml of an aqueous containing 0.5473 g La(NO₃)₃.6H₂O and 0.0288 g Er(NO₃)₃.5H₂O was added drop wise, while stirring the solution continuously. The reaction was allowed to continue for two hours and then slowly cooled to room temperature. Around 100 ml of 100% ethanol was added to the reaction medium to precipitate the citrate stabibilzied LaF₃:Er particles. The particles were separated by centrifugation at 3500 rpm, washed with ethanol, and dried under vacuum.

Synthesis of Indium sulfide nanoparticles doped with Eu³⁺ (In₂S₃:Eu)

1.74 gms of (C₁₈H₃₇O)₂PS₂ NH₄⁺ and 0.054 gms of sodium sulfide was dissolved in 130 ml of ethanol: water (1:1) mixture at 75 °C. A solution of indium chloride (50 mg) and europium chloride hexahydrate (4 mg) in 2 ml of water was added drop-wise and the solution was stirred at 75 °C for 3 hrs and then cooled to room temperature. Ethanol 50 (ml) was added and the mixture was stirred for 5 min. Subsequently, the yellow-coloured precipitate was centrifuged and washed with ethanol. The

nanoparticles were dried in vacuum for two days. After drying the nanoparticles were dispersible in chloroform and dichloromethane.

Preparation of SiO₂ thin films incorporated with LaF₃:Er particles.

Around 50 mg of LaF₃:Er-citrate particles were dissolved in 1.0 ml water. 3ml of TEOS (tetraethoxysilane) and 7.8 ml of ethanol were added to this, followed by few drops of concentrated HCl till the pH reaches around 2. The TEOS: ethanol: water ratio was maintained around 1:10:8.

The resulting solution was stirred for 2 days at room temperature to get a viscous solution which is then spin coated on a quarts substrate. The spin coated samples were heated at 800°C for 12 hours.

Waveguide fabrication. The PMMA/ LaF₃:Nd solution was spincoated on a 3" silicon wafer coated with 3.2 μm SiO₂ (Plasma enhanced chemical vapor deposition PECVD) on top. The solution consisted of 900 mg PMMA and 100 mg of nanoparticles dissolved in 4 g of chlorobenzene. Spincoating was performed at 1250 rpm resulting in a layer thickness of 3.3 μm. After spincoating the wafer was dried at 95 °C followed by spincoating a 1.2 μm thick layer of positive photoresist on top. Patterns were made with a mask having channels ranging in diameter from 2-10 μm, using a radiation time of 3.7 s. The photoresist was developed in standard OPD developer for 50 sec. After creation of the pattern in the photoresist the polymer was etched using reactive ion etching (RIE). RIE was done for 8 min at a power of 20 W with a pressure of 10 mTorr and 20 sccm O₂, keeping the wafer at a temperature of 10

 $^{\circ}$ C. Excess photoresist was removed by irradiation and development, leaving ridges with a height of 1.45 μm .

The doped SU-8 layers were spincoated on 10 cm borosilicate wafers that were previously cleaned in fuming nitric acid for 8 minutes. The solid content of the solution of 52 % included the nanoparticles and the composition of the SU-8 solution in a 1:4 weight ratio. Spincoating was done at 1250 rpm for 30 seconds with an acceleration of 100 rpm/sec. The layers were dried at 65 °C for two hours, followed by exposure for 300 seconds through a mask containing the waveguide structures. The film was postbaked at 65 °C for 2 min followed by development in RER 600 developer for 2 minutes. The structures were washed with isopropanol to remove excess developer and dried with a nitrogen stream. Flood exposure was done for 5 minutes, followed by a hard bake for 2 hours at 150 °C. A cladding of HEMA-styrene was spincoated on top of the device at a spin speed of 750 rpm.

Photophysical measurements. Measurements on the waveguides were performed using an Ar⁺ pumped dye laser as the pump laser at 578 nm and a TiSapphire laser was used for pumping at 795 nm. The signal laser was a diode laser operating at 1319 nm. The laser light was coupled into a 9/125 µm single mode fiber using a microscope objective and from the fiber into the waveguide. Alignment of the fibers and chip were done using a butt-end coupling setup. Light coming from the waveguide was collected with a single or multi-mode fiber and lead to a detector or a spectral analyzer for emission spectra.

Example 1:

Characterization of Eu³⁺ doped nanoparticles. The nanoparticles of TiO₂:Eu were characterized by TEM, by evaporating a drop of a nanoparticle solution on a carbon coated copper grid. A typical picture of nanoparticles obtained by heating the reaction mixture for 30 min is shown in Figure 1. The picture shows small particles with low contrast, due to the small size of the nanoparticles and the low atomic weight of the Ti and O atoms. The nanoparticles were between 3-5 nm in size and in some nanoparticles lattice fringes can be seen, indicating that they are crystalline. The inset shows an enlargement of such a crystalline nanoparticle in which the lattice planes are more clearly observed. The size of the nanoparticles was similar to the sizes found for un-doped TiO₂ nanoparticles synthesized using the same method.

The composition of the nanoparticles was measured using X-ray fluorescence (XRF). For a sample synthesized with a molar ratio of Eu³⁺ to Ti⁴⁺ of 1:17.7 and a reaction time of 30 minutes, a weight ratio of Eu₂O₃:TiO₂ 10.9:61.6 % was found, giving a molar ratio of 1:12.5 in the nanoparticles. The molar ratio of Eu³⁺ in the nanoparticles was higher than applied in the synthesis, which might have be a result of incomplete precursor decomposition at this reaction time. The remaining weight consisted of the TOPO capping groups around the nanoparticles.

The emission of the Eu³⁺ ion was used as a probe for the environment around the lanthanide ion. The sensitivity of the emission lines of Eu³⁺ to the symmetry of the surrounding gives valuable information on the site in which the ion is located and it gives the possibility to compare the emission spectrum with emission spectra from the literature of Eu³⁺ doped in other TiO₂ materials. The emission, absorption and

excitation spectra of TiO_2 :Eu nanoparticles in dichloromethane are shown in Figure 2. The emission spectrum showed typical Eu^{3+} luminescence from the 5D_0 level after excitation in the TiO_2 absorption band.

The emission spectrum showed a striking similarity to the emission spectra of Eu^{3+} in a TiO_2 thin film and in mesoporous TiO_2 . The ratios of the different peaks in the emission spectra were the same and also the peak splitting is very similar, which was a strong indication that the environment around the Eu^{3+} ion is the same. The emission of the $^5D_0 \rightarrow ^7F_2$ transition was clearly the dominating emission line, indicating the Eu^{3+} ion was in an environment without inversion symmetry. The peaks of the transitions were not well resolved. Sharp transition lines were observed when the Eu^{3+} ions were all doped in a limited amount of well-defined crystal sites. The broad emission lines found in the emission spectrum of the TiO_2 :Eu nanoparticles could have been caused by the doping of Eu^{3+} in amorphous material, but in the TEM pictures crystalline nanoparticles were found. Another reason for the broadening of the emission lines can be the distortion of the crystal lattice around the Eu^{3+} ion. The size and charge mismatch between the two cations (Eu^{3+} 0.95 Å, Ti^{4+} 0.69 Å) leads to oxygen vacancies around the Eu^{3+} ions. This makes the sites of each of the Eu^{3+} ions slightly different, leading to broadening of the emission lines.

The absorption spectrum of the doped nanoparticle solution is also shown in Figure 2. A strong absorption band of the TiO₂ host is observed starting at 350 nm. The onset of this absorption is the band gap of the TiO₂ host and this is blue shifted, compared to bulk TiO₂, with a band gap of 382 nm. The blue shift of the band gap is caused by quantum size effects, due to small size of the nanoparticles. The lanthanide

absorption bands are not observed in the absorption spectrum, because of the low absorption coefficients.

The excitation spectrum of the Eu³⁺ emission at 610 nm is very similar to the absorption spectrum, except that the absorption peaks of the Eu³⁺ ion are observed at 395 and 465 nm. The presence of the TiO₂ absorption band in the excitation spectrum demonstrates the energy transfer from the semi-conductor host to the Eu³⁺ ion.

The luminescence decay of the TiO₂ nanoparticles after excitation at 300 nm is shown in Figure 3. Energy transfer from the semi-conductor host to the lanthanide ion is very fast, because in the decay no build-up in the luminescence is observed.

The decay shows a deviation from mono-exponential behavior, just like for the other lanthanide doped nanoparticles described herein. Fitting the decay with a bi-exponential fit gave the luminescence lifetimes of the Eu³⁺ emission. A long component of 1.2 ms responsible for 77 % of the luminescence and a short component of 0.4 ms responsible for 23 % of the luminescence was found. The equation of Werts et al incorporated herein by reference (M.H.V. Werts, R.T.F. Jukes, J.W. Verhoeven Phys. Chem. Chem. Phys. 2002, 4, 1542), was used to calculate the radiative lifetime of the Eu³⁺ ion and to make an estimate of the intrinsic quantum yield of the Eu³⁺ ion. Using the refractive index of dichloromethane, a radiative lifetime of 2.9 ms was calculated from the shape of the emission spectrum. This radiative lifetime implies that the intrinsic quantum yield of the Eu³⁺ ion is about 40%. The effective refractive index that the Eu³⁺ ions experience is not exactly known. The ions itself are doped in the TiO₂ host material with a refractive index of

2.6-2.9 and the nanoparticles are dissolved in dichloromethane with a refractive index of 1.42. It is therefore likely that the refractive index the Eu^{3+} ions experience is a bit higher than the refractive index of the solvent and this would lead to a shorter radiative lifetime and as a consequence a slightly higher intrinsic quantum yield. The luminescence lifetime of the Eu^{3+} ion in these nanoparticles is higher than the luminescence lifetime of Eu^{3+} in mesoporous TiO_2 of 500 μ s. This difference can be attributed to the influence of the solvent with the lower refractive index, making the radiative lifetime of the Eu^{3+} ion longer.

A quantum yield of the sensitized emission of the TiO₂:Eu nanoparticles was determined by comparing the emission intensity with the emission intensity of a solution of ruthenium tris bipyridine in deoxygenated water with the same optical density at the excitation wavelength. The determined quantum yield of the sensitized Eu³⁺ emission was only 0.03 %. This very low quantum yield is in contradiction with the high intrinsic quantum yield of the Eu³⁺ ion.

Example 2:

Quenching in lanthanide doped TiO_2 nanoparticles. The quenching routes of the lanthanide ion in the TiO_2 nanoparticles were investigated to explain the low quantum yield. Residual OH groups at the surface of the nanoparticles or in the core of the nanoparticles can quench the luminescence of the lanthanide ion. An FT-IR spectrum was measured to see if water or OH groups are present in the product (Figure 4). The nanoparticles were dried over P_2O_5 to remove any physisorbed water present on the nanoparticles surface after the synthesis.

A broad band centered at 3300 cm⁻¹ was found, which can be ascribed to the OH stretching vibration. The peaks at 2925 cm⁻¹ and 2850 cm⁻¹ are due to stretching vibrations of the CH bonds of the TOPO ligand, together with the PO band around 1046 cm⁻¹. The band around 500 cm⁻¹ is caused by TiO bonds in the TiO₂ host. To investigate the effect of quenching by OH impurities, the nanoparticles were also synthesized by using deuterated solvents. Nanoparticles were synthesized by dissolving the EuCl₃·6H₂O in MeOD and washing the product with MeOD after the reaction. This procedure will change free OH bonds inside and on the surface of the nanoparticles in OD bonds and thereby reduce the quenching of the Eu³⁺ ion. When the luminescence intensity and the luminescence lifetimes of the Eu³⁺ ion in nanoparticles synthesized with methanol and with methanol- d_1 were compared, no change was found, indicating that OH bonds are present, but not as an important source of quenching. To increase the quantum yield of the nanoparticles, the influence of the reaction time was investigated. A longer reaction time could influence the quantum yield by increasing the crystallinity, improving the surface capping, and by a growth of the nanoparticles. In Figure 5 the luminescence lifetime of Eu³⁺ is plotted as a function of the reaction time and an increase in lifetime is clearly observed.

This increase in the luminescence lifetime suggests an increase in quantum yield for a longer reaction time, but this was not found. It is possible that un-doped nanoparticles are formed at longer reaction time, which also absorb light, but do not contribute to the lanthanide luminescence. In this case the quantum yield of the doped nanoparticles is increased, but the result of the quantum yield measurement is disturbed by the presence of un-doped nanoparticles. A change in symmetry around

the Eu³⁺ ion could also lead to an increase in the measured lifetime, but this symmetry change should also lead to a different emission spectrum and this was not observed. The relatively long lifetime of the Eu³⁺ ion is an indication that once the energy is transferred to the lanthanide ion, quenching of the Eu³⁺ ion is not the dominating process. Thus, the overall quantum yield of the particles should be improved by decreasing the non-radiative decay of the TiO₂ host.

Example 3:

Lanthanide ions emitting in the Near Infrared. Other lanthanide ions were also successfully doped in the TiO₂ nanoparticles. The emission and excitation spectra of TiO₂ nanoparticles doped with the NIR emitting lanthanide ions Nd³⁺, Er³⁺, and Yb³⁺ are shown in Figure 6.

After excitation above the TiO_2 band gap, the emission spectra show the typical luminescence of these lanthanide ions. The Nd^{3+} ion emits at 880, 1060, and 1330 nm originating from the ${}^4F_{3/2} \rightarrow {}^4I_J$ (J = 9/2, 11/2, 13/2) transitions, respectively. The Er^{3+} ion emits at 1530 nm from the ${}^4I_{13/2} \rightarrow {}^4I_{15/2}$ transition and the Yb^{3+} ion at 980 nm from the ${}^2F_{5/2} \rightarrow {}^2F_{7/2}$ transition. The excitation spectra show the same broad band starting at 350 nm the same as for the TiO_2 : Eu^{3+} particles proving that sensitized emission of these ions also occurs. At high particle concentrations of 50 mg/ml, the excitation spectra also show the direct excitation of Er^{3+} and Nd^{3+} , because the lanthanide absorption lines become visible in the excitation spectra (Figure 6). In the high concentration samples the broad band decreases below 310 nm, because at these concentrations all the excitation light is absorbed in the first millimeters of the fluorescence cuvet and the emission is collected at an angle of 90 degrees from the

middle of the cuvet. The emission of the Nd³⁺ and Er³⁺ ions around 1300 and 1500 nm are of special interest for telecommunication purposes, because these wavelengths are in the telecommunication windows.

Example 4:

LED with TiO₂:Eu nanoparticles. An LED doped with TiO₂:Eu nanoparticles was fabricated, with a structure as shown in Figure 7.

On the ITO covered glass slides a layer of 80 nm of PEDOT was spin-coated to improve the hole injection from the ITO into the device. When a nanoparticles solution was spin-coated on top of the PEDOT layer cracks appeared in the nanoparticle layer, which can lead to shortcuts in the LED. For this reason a mixture of nanoparticle/PMMA 3/1 was used, because smoother layers could be spin-coated using the polymer blend. Although the PMMA polymer is a non-conducting polymer a semi-conducting layer can be fabricated when the doping ratio of the nanoparticles is high enough. Electroluminescent devices have been fabricated of nanoparticle PMMA blends using ZnS:Cu nanoparticles, which have a bit larger band gap than our TiO₂ nanoparticles. A drop of InGa eutectic was used as the cathode. The device behaved as a diode, which can be seen by the current-voltage characteristics as shown in Figure 8. The turn on voltage of the device is around 12 V. Up to a voltage of 21 V when a significant current is going through the device no emission could be detected. The low quantum yield of the nanoparticles that was found in the photo-physical measurements is one of the reasons why it will be very difficult to observe electroluminescence.

Example 5:

Typical pictures for LaF₃:Ln and LaPO₄:Ln (Ln = Eu^{3+} and Nd^{3+}) are shown in Figure 9.

The morphologies of both particles are independent of the dopant ion. The high crystallinity of the particles can clearly be seen, despite the fact that the particles were synthesized at relatively low temperatures, i.e. below 200°C. The LaPO₄ particles with an average size of 4.3 nm are somewhat smaller in size than the LaF₃ particles with an average size of 6.3 nm. In addition, the size distribution of 15% for the LaPO₄ particles is lower than the size distribution of the LaF₃ particles, which is 28%. The LaF₃ particles are less spherical in shape with aspect ratios between 1 and 3. The spherical and elongated particles both consist of one crystal domain, so the elongated particles are not formed by the aggregation of multiple particles. In the size distribution histograms the sizes of the elongated particles were determined by averaging of the long and the short axis. The sizes of both particles are sufficiently small to prevent scattering when the particles are incorporated in a polymer matrix.

The elemental composition of the particles was determined using X-ray fluorescence (XRF) and combustion elemental analysis. The results are summarized in Table 1.

Table 1: Elemental composition of Eu³⁺-doped nanoparticles (weight %)

Particle	La ^a	Ln ^a	P ^a	C_p	H^b	S^b	N^b
LaF ₃ :Nd	24.76	1.23	3.48	42.49	7.76	5.76	••
LaF ₃ :Er	24.20	1.35	3.48	43.59	7.62	6.48	-
LaF ₃ :Ho	35.41	1.94	2.94	33.07	7.76	5.72	-
LaPO ₄ :Nd	41.97	2.42	12.19	6.33	2.16	. -	1.01
LaPO ₄ :Er	41.80	2.36	12.56	5.57	2.63	-	1.20
LaPO ₄ :Pr	41.97	2.35	12.52	6.30	2.17	-	1.01

a) measured with XRF

c) n. d. is not determined

The same ratio of lanthanide ions as applied in the synthesis was found in the particles reflecting the similar reactivity of the lanthanide ions. The rest consists of organic groups on the outside of the particles like tetramethyl ammonium ions and ethylhexyl chains for the LaPO₄ particles and the coordinated dithiophosphate ligands for the LaF₃ particles. The surface of the LaPO₄ nanoparticles consist of deprotonated phosphate groups coordinated to ammonium ions and some of the solvent and hydrolyzed products of the solvent. The coordination of the organic molecules can be seen from the broadened signals in the 1H NMR and the ratio of Ln/P, that is smaller than one, indicates that phosphate groups are at the surface and not the lanthanide ions. For the LaF₃ particles it is likely that the surface consists of Ln³⁺ ions, because these can coordinate to the negatively charged ligand that stabilizes the particle. The coordination of the ligand to the surface of the particles can be seen from the

 \hat{Q}

b) measured with combustion elemental analysis.

broadened signals of the ligand in the ¹H NMR spectrum. Fluoride could not be detected using both these analysis techniques so no information could be obtained about the amount of F in the LaF₃ particles. The amount of ligand measured with elemental analysis on the LaF₃ particles is not always constant. For the Nd³⁺ and Er³⁺ doped particles the amount of carbon and hydrogen measured is the same, but the amount of sulfur is slightly lower for the Nd³⁺ particles. A deviation in the amount of sulfur could be caused by oxidation of the ligand after prolonged exposure to air, as found by mass spectrometry. This oxidation leads to a ligand with a phosphate headgroup, which binds more strongly to the particles and would therefore lead to even more stable particles. For the Ho³⁺ doped particles the amount of carbon and hydrogen is lower than for the other two ions. The particle size and size distribution is the same as for the other two batches, so apparently there is less ligand coordinated to the Ho³⁺ doped particles. For all these particles no free ligand is present because no nitrogen was detected and because only broadened signals are present in the ¹H NMR spectra proving the coordination of the ligand to the particle surface. properties of nanoparticles usually play an important role in the luminescence properties and it is therefore important to know which molecules are coordinated to the surface.

Example 5:

Luminescence of Eu^{3+} doped particles. The Eu^{3+} ion was used as a probe for the crystal environment in which the ion is located. Typical Eu^{3+} emission originating from the 5D_0 and 5D_1 levels was observed after direct excitation of the ion at 397 nm for LaF₃:Eu and 260 nm for LaPO₄:Eu (Figure 10).

In LaPO₄ the Eu³⁺ can be excited at 260 nm because of a charge transfer band. This charge transfer band is visible in more phosphors in which Eu3+ is coordinated to oxygen and is caused by an electron transfer in the Eu³⁺-O²⁻ bond. Deactivation of the charge transfer state leads to an excited Eu³⁺ ion. Due to the low reduction potential of Eu³⁺ this band occurs at wavelengths above 200 nm. In the emission spectrum the ratio of the different peaks of the $^5D_0 \rightarrow ^7F_J$ (J = 1, 2) transition gives information about the symmetry of the crystal site in which the ion is located. This confirms that the dopant ion is present in a La³⁺ crystal site giving the Eu³⁺ ion C₁ symmetry in LaPO₄ and C₂ symmetry in LaF₃ particles, the same as for the bulk materials. In the LaF₃:Eu particles emission from the ⁵D₁ level at 524, 534, 554, and 582 nm can clearly be observed as well. Emission from this level also occurs in the LaPO₄:Eu particles only with much less intensity. This large difference in emission from the ⁵D₁ level is caused by the difference in phonon energies of the host materials. The ⁵D₁ level lies only 1750 cm⁻¹ above the ⁵D₀ level and thus the ⁵D₁ level can be depopulated very fast to the ⁵D₀ level by multi-phonon emission in LaPO₄. In the LaF₃ host the maximum phonon energies are so low that multi-phonon emission becomes inefficient. The different symmetry sites of Eu3+ in these particles can clearly be observed from the different emission spectra. The Eu³⁺ ion can also be used as a probe for more similar crystal structures as is illustrated in LnPO₄ nanoparticles with Ln = La, Gd, Y, and Lu, synthesized using the same procedure described for LaPO₄ particles. The crystal structure of these phosphates is dependent on the size of the lanthanide ion. A monazite type is found for the larger lanthanide ions (Ln = La^{3+} - Gd^{3+}) and xenotime for the smaller lanthanide ions (Ln = Tb^{3+} - Lu^{3+}). The excitation and emission spectra for LnPO₄:Eu (Ln = La³⁺, Gd³⁺, Y³⁺, Lu³⁺) nanoparticles are shown in Figure 11.

The excitation spectra are scaled to the 393 nm peak of Eu³⁺ and all show a charge transfer band, but the excitation maximum is blue shifted upon a decrease in ion size. The excitation spectra of LuPO₄ and YPO₄ exhibit a sharp cut-off around 220 nm, which is caused by the absorption of the excitation light by the solvent. The blue shift of the charge transfer band upon decreasing the ion size can be explained by a stronger binding of the phosphate groups to the smaller trivalent ions and by the smaller crystal site the Eu³⁺ ion has in the LuPO₄ and YPO₄ lattices. This makes the phosphates more difficult to oxidize and the reduction of Eu³⁺ to Eu²⁺ less favorable. The sharp lines in the GdPO₄ host at 273, 276, 278 nm and some smaller peaks around 310 nm are due to the excitation of Gd^{3+} ($^8S_{7/2} \rightarrow ^6I_J$, 6P_J , J = 3/2, 5/2, and 7/2) and subsequent energy transfer to the Eu³⁺ excited states. The emission spectra of all four LnPO₄:Eu materials exhibit the characteristic Eu³⁺ emission lines for the $^5\mathrm{D}_0 \!\!\to^7\!\! \mathrm{F}_J \, (J=0\text{-}4)$ transitions. Some differences are observed in the emission spectra of the two host materials with the larger ions (La³⁺ and Gd³⁺) compared to the host materials with the smaller ions (Y3+ and Lu3+). The relative intensities of the $^5D_0 \rightarrow ^7F_2$ transition are different: 0.30, 0.30, 0.43, 0.50 for the doped LaPO₄, GdPO₄, LuPO₄, and YPO₄, respectively. These values correspond to a crystal site with a high degree of inversion symmetry and more inversion symmetry for the La and Gd phosphates than for the Lu and Y phosphates. Furthermore, a number of extra peaks are found in the $^5D_0 \rightarrow ^7F_4$ transition, all consistent with the difference in crystal structure between the LaPO₄, GdPO₄ and the LuPO₄ and YPO₄ particles.

The luminescence decay of the ion gives information on the influence of quenching from the organic solvent and ligand outside the particle and quenching by impurities

from inside the particle. An example of the decays of LaF₃:Eu and LaPO₄:Eu particles is given in Figure 12.

Excitation occurs in the higher lying Eu³⁺ levels followed by a rapid decay to the luminescent ⁵D₀ level. For the LaPO₄:Eu particles, the luminescence of the ⁵D₀ level shows a short rise with a time constant of 24 μs . This is similar to the decay time of the ⁵D₁ level of 23 μs showing that the ⁵D₀ level is partially populated through the ⁵D₁ level. After this short rise in the LaPO₄:Eu luminescence, both decays are clearly not mono-exponential, so not all ions in the particles have the same probability of decay. An acceptable fit was obtained when two exponentials were used. However, there is no physical reason to use just two exponentials. Quenching by groups close or at the surface of the nanoparticle should lead to a gradual change of quenching efficiency of the lanthanide ion from the surface to the core of the particle. Assuming a random distribution of the dopant lanthanide ion over the nanoparticle and that all ions are in the same crystal site, there is no physical justification for fitting the decay curve with just two exponentials, which suggests two populations. A random distribution of the doping ions is reasonable because of the high similarity in cation size, charge, and reactivity between the lanthanide ions. It was possible to check if all the ions are in the same crystal site. First of all, the emission of the $^5D_0 \rightarrow ^7F_0$ transition gives information on the amount of crystal sites the ion is doped in. This is a transition between two non-degenerate levels, so if all the ions are in the same crystal site, only one emission peak for this transition is present around 578 nm. In both Eu³⁺ doped particles of LaF3 and LaPO4 we indeed see just one peak for this transition, measured at a resolution of 0.1 nm (inset Figure 10), indicating that all ions are in the same crystal site. Time-resolved spectroscopy is another method giving information on

different crystal sites. Ions doped in different crystal sites have a different symmetry leading to a difference in emission spectrum and a difference in luminescent lifetime. An emission spectrum taken directly after an excitation pulse is caused by emission from both types of ions, but an emission spectrum at the end of the decay is the emission from ions with the longest lifetime. When all ions are in the same crystal site no differences in emission spectrum should be observed. Figure 13 shows the emission spectra of LaPO₄:Eu in methanol 0.2 ms after an excitation pulse and 29 ms after an excitation pulse.

Although the intensity at the end of the decay is much lower, the shape of the emission spectrum is essentially the same as in the early stages of the decay, thereby confirming that Eu³⁺ is doped in one single crystal site.

Example 7:

Modeling the luminescence decay. When it is assumed that most of the quenching of the excited state is located close to the surface, a model can be designed taking into account these surface effects. Ions in the core of the particles will have the longest lifetime and going to the outside the lifetime will decrease. When a particle is divided into a number of shells, the lifetime of one shell can be related to the lifetime in another shell by an effective radiative lifetime and a quenching factor that is dependent on the distance to the surface. The luminescence decay can thus be described using the following equation:

$$I_t = I_0 \sum_{i=1}^n \frac{1}{n} e^{-k_i t}; k_i = \frac{1}{\tau_i} = k_R + C \times f_{q,i}$$

In this equation I_t is the luminescence intensity at time t, I_0 the intensity at t = 0, n the number of shells, $k_i = 1/\tau_i$ the rate constant in shell I, with τ_i the luminescent lifetime of this shell, t the time, $k_R = 1/\tau_R$ a rate constant in the absence of quenching from outside the particle so it is an apparent radiative decay rate, and C the quenching constant. k_R is an apparent decay rate, because it only takes into account quenching from the surface. All other quenching processes inside the nanoparticle, like concentration quenching and quenching by impurities inside the nanoparticle can also be present. The constant C is a fit parameter that is dependent on the individual ion, the size and size distribution of the particles, and on the strength of the quenching. The size and size distribution are still in the quenching constant, because at this moment no size dependent decays are available. The relative quenching factor fq,i takes into account the distance of a shell to the surface and the distance dependence of When quenching is assumed to take place via a dipole-dipole the quenching. mechanism, the quenching has a distance dependence of r⁻⁶. The total quenching then has to be integrated over the total volume outside the particle. Integration over the total volume gives the total quenching factor $f_{q,l}$, as shown in the following equation:

$$f_{q,i}(a_i) = \int_{V} r^{-6} dV$$

This equation was solved for a particle shell i, with $a_i=A_i/R_i$ as the center of each shell, by varying the radius r (Figure 14). Three domains can be defined. The first domain is totally inside the particle ranging from r=0 to r=(R-A), so for this domain no quenching is taken into account. In the second domain the volume fraction $d[V(r,\phi)]$ is dependent on both the radius and the the angle ϕ (part of the shell is inside the particle with no quenching and part is outside the shell, but with quenching). This domain ranges from r=(R-A) to r=(R+A) (Figure 14).

The third domain lies totally out of the particle and is from r = (R+A) to ∞ . The fraction of a shell in the second domain that is outside the particle equals $2\pi r^2$ (1- $\cos \varphi$)dr. The volume fraction dV of a shell in the third domain is $4\pi r^2$ dr. Multiplying this with r^{-6} results in the following integral, that was numerically solved for a particle consisting of 10 shells of equal volume(see equation below). Assuming that the ions are randomly distributed in the particle will give all ions the same weight in the fit.

$$f_{q,i}(A,\varphi) = \int_{V} r^{-6} dV = \int_{R-A}^{R+A} \frac{1}{r^{6}} 2\pi r^{2} (1 - \cos\varphi) dr + \int_{R+A}^{\infty} \frac{1}{r^{6}} 4\pi r^{2} dr$$

This results in relative quenching factors for the different shells as shown in Table 2

Table 2: Quenching factor $f_{q,i}$ for 10 shells

		·
Shell i	a_i	$f_{q,i}(a_i)$
1	0.23	4.9
2	0.52	11
3	0.63	19
4	0.70	32
5	0.77	63
6	0.82	121
7	0.87	$3.0\ 10^2$
8	0.91	$8.6\ 10^2$
9	0.95	$4.9\ 10^3$
10	0.98	7.7 10 ⁴

Using these values for the quenching factors, the decays of the particles can be fitted by varying the values for the apparent radiative lifetime τ_R and C. Good fits were obtained for LaPO₄:Eu and LaF₃:Eu particles (Figure 12, solid lines). In these fits the luminescence of the 5D_1 level is not taken into account. Emission from this level contributes only very little to the total emission, but it could have a small influence on the obtained values.

For LaPO₄:Eu particles in methanol, the value for τ_R is 7.0 ms with a quenching constant of 4.1 10^{-2} s⁻¹. In the case of LaF₃:Eu particles in dichloromethane, the τ_R is 7.7 ms with a quenching constant of 2.9 10⁻¹ s⁻¹. These results show that the apparent radiative lifetime of Eu³⁺ in LaF₃ is longer than in LaPO₄ conform the literature, but that the ion is quenched more strongly in the LaF3 particles than in the LaPO4 particles. The difference in solvent does not explain the difference in quenching strength, because methanol is a stronger quenching solvent than dichloromethane due to the presence of an O-H bond. The different termination of the surface can be the reason for a different quenching strength. The LaPO₄ particles are terminated with phosphate groups, but the LaF₃ particles are terminated with Ln³⁺ groups. This means that in the case of LaF₃ particles the luminescent ion can directly coordinate to a quenching group of the organic environment or the ligand, which is not the case for the LaPO₄ particles. One other possible source of quenching is the coordination of water to the surface of the particles. It has been shown that the lifetime of LaF₃:Eu particles is longer when the particles are synthesized in deuterated solvents proving that residual OH groups are present after the synthesis and act as quenchers. When this product was stirred in a water/methanol mixture the lifetime of Eu3+ went back to

the value for particles synthesized in non-deuterated solvent, proving that the OH groups are at the surface of the particles and accessible to the solvent. The LaPO₄ particles are synthesized under near water free conditions and any water present will be used in the hydrolysis of the solvent so contamination with water is probably less important. The apparent radiative lifetimes ($\tau_R=1/k_R$) do not necessarily relate to the real radiative lifetime, because quenching inside the particle is not taken into account in this model. However, the high values do indicate that quenching inside the particle is not significant. The size dispersity of the particles is in the quenching constant C, so the values of C for LaF₃ and LaPO₄ particles cannot be compared directly with each other. Figure 12 shows the variation of the lifetime τ_1 in the different particle shells. As expected the longest lifetimes are observed in the core of the particles and going to the surface of the particles the lifetime decreases consistent with quenching by the solvent. Because the total luminescence is the area under the decay curve, an average lifetime of these ten values can be calculated using the following equation:

$$\tau_{\rm av} = \frac{\sum A_i \tau_i^2}{\sum A_i \tau_i}$$

For LaF₃:Eu an average lifetime of 6.5 ms is calculated and for LaPO₄:Eu particles an average lifetime of 6.6 ms. The values for these materials are close to the radiative lifetimes calculated and measured previously for the bulk materials, but unfortunately it is not possible to compare them directly to these radiative lifetimes, because the nanoparticles are in another solvent. However, they do indicate a high quantum yield.

Example 8:

Quantum yield. A quantum yield of these lanthanide-doped particles can be determined using Ce³⁺ doped particles.¹ Ce³⁺ has a high absorption between 200-300 nm because of an allowed 4f-5d transition. It is known that Ce³⁺ shows efficient energy transfer to Tb³⁺, which emits in the visible and from particles doped with both these ions it is possible to determine the quantum yield by comparison with a dye. We determined the quantum yield by using quinine bisulfate in 1M H₂SO₄ as standard. For La_{0.4}Ce_{0.45}PO₄Tb_{0.15} a quantum yield of 10% in methanol was found, lower than the value of 24% found by Haase and co-workers. The value found by Haase et al. herein incorporated by reference (K. Riwotzki, H. Meyssamy, H. Schnablegger, A. Kornowski, M. Haase, Angew. Chem. Int. Ed. 2001, 40, 573), was measured using Rodamine 6G and when we used this dye a quantum yield of 21 % was found, which is within the experimental error of about 2% (absolute). The value found using quinine bisulfate is probably more reliable, because this dye is more suitable for excitation in the UV. The Rodamine 6G dye has a low absorption at the excitation wavelength and much overlap between emission and absorption so at higher concentration re-absorption of emission could occur leading to an overestimation of the quantum yield. The emission and absorption of quinine bisulfate have very little overlap so self-quenching at higher concentrations does not take place. It was also possible to synthesize La_{0.4}Ce_{0.45}F₃Tb_{0.15} particles and the emission spectrum is shown in Figure 15 together with the emission spectrum of quinine bisulfate.

For La_{0.4}Ce_{0.45}F₃Tb_{0.15} in dichloromethane a quantum yield of 19% was found. When D₂O was added to a dichloromethane solution of LaCeF₃Tb particles an increase in quantum yield to 38% was observed. This confirms the coordination of a small amount of water to the surface of the particles leading to quenching of the excited state of the lanthanide ions. The high-energy vibrations of water molecules effectively quench the excited state of the lanthanide ions and this can be reduced by an exchange with deuterium because of the lower vibrational energy of the OD bond. This coordination of water is in full agreement with previous results where particles of LaF₃:Eu were synthesized in deuterated solvents, leading to an increase in the luminescence lifetime.

Example 9:

Near-infrared emission. Particles of LaPO₄ and LaF₃ were also doped with the NIR emitting ions Nd³⁺, Er³⁺, Pr³⁺, and Ho³⁺. Luminescence spectra of these ions are shown in Figure 16.

The luminescence spectra of Nd^{3+} in both particles show the typical emission at 880, 1060 and 1330 of the ${}^4F_{3/2} \rightarrow {}^4I_{13/2}$, ${}^4F_{3/2} \rightarrow {}^4I_{11/2}$ and ${}^4F_{3/2} \rightarrow {}^4I_{9/2}$ transitions, respectively The emission spectrum measured for LaF₃:Nd looks very similar to the spectrum measured in a Nd^{3+} doped LaF₃ thin film. The emissions are almost the same for both types of particles, showing only slightly different splitting of the emission peaks, indicative of a different symmetry around the luminescent ion, similar to that found for the Eu^{3+} ion. However, the differences in the luminescence spectra are smaller than for Eu^{3+} , because Eu^{3+} is much more sensitive to the symmetry of the crystal site than the other lanthanide ions. The excitation spectra for both particles also look very

similar and are comparable with the absorption lines of the Nd³⁺ ion in solution. The wavelength positions of the peaks do not change, only the peak splitting and the ratio between peak intensities differ. Er³⁺-doped materials show typical luminescence at 1530 nm of the ${}^4I_{13/2} \rightarrow {}^4I_{15/2}$ transition in both particles with a small difference in the shape and the width of the emission peak caused by a difference in crystal symmetry. The emission of the LaF₃:Er particles again looks very similar to the emission spectrum of an Er³⁺-doped LaF₃ thin film. For the LaPO₄ host, a small emission is observed at 980 nm of the ${}^4I_{11/2} \rightarrow {}^4I_{15/2}$ transition, which is not visible in the LaF₃ host, probably because of symmetry reasons. The excitation spectra are very similar for both particles, showing small differences in peak intensities caused by symmetry differences. For the Pr³⁺ ion, luminescence can originate from different levels depending on the crystal field and on the quenching mechanisms. Luminescence originating from the ¹D₂ level was observed in LaPO₄:Pr particles by excitation with the 476 nm line of an Ar⁺ laser. When the LaF₃:Pr particles were excited using the same laser line, no emission was observed, although it is known that the Pr³⁺ ion can emit very efficiently in this host matrix. It is possible that Pr³⁺ does not absorb at this wavelength in the LaF₃ host or that the Pr³⁺ ion is very sensitive to the small amount of water present at the surface of the LaF₃. Under excitation with an Xe arc lamp at 443 nm, the emission of LaPO₄:Pr could also be detected, but only weak emission was observed for LaF₃:Pr particles. For the Ho³⁺ ion, emission could be detected in the LaF₃:Ho particles at 960 and 1460 nm from the ${}^5F_5 \rightarrow {}^5I_7$ and ${}^5F_5 \rightarrow {}^5I_6$ transitions. In the visible, an emission of the ${}^5F_5 \rightarrow {}^5I_8$ transition at 640 nm was also observed. In this host, Ho³⁺ luminescence is known to occur from a variety of levels. The dominant luminescence found in the bulk material is from the ⁵S₂ level at 540 nm, which is, however, not observed in our nanoparticles. A reason for this can be that this level

lies approximately 3000 cm⁻¹ above the ⁵F₅ level which corresponds to the vibrational energy of a CH bond. The presence of the ligands on the surface of the particle could lead to a fast non-radiative decay of the ⁵S₂ level to the ⁵F₅ level from which luminescence occurs. The LaPO₄:Ho particles did not show any luminescence at all. Ho³⁺ emission in LaPO₄ crystals is not reported in the literature probably because the phonon energies in LaPO₄ are too high, leading to relatively efficient non-radiative processes. NIR emission of Pr³⁺ and Ho³⁺ in an organic environment has not been reported until now, proving the good shielding against the organic groups when they are doped in the nanoparticles. Using these ions the spectral window between 1300-1600 nm, which is of interest to telecommunication purposes, can be covered completely.

Example 10:

Yb³⁺ as a sensitizer. To increase the effective absorption of the Er³⁺ ion, co-doping with Yb³⁺ is often used. Yb³⁺ has a higher absorption at 980 nm than Er³⁺ and it is able to transfer energy to the Er³⁺ ion. The 980 nm wavelength is commonly used for pumping Er³⁺ ions because at this wavelength a large variety of cheap semi-conductor pump lasers are available. Yb³⁺ also has a low reduction potential leading to charge transfer bands in phosphates, similar to Eu³⁺, so the possibility of a charge transfer in LaPO₄ particles was investigated. The emission and excitation spectra of a sample of LaPO₄:Yb in methanol is shown in Figure 17.

In the excitation spectrum a broad band peaking at 250 nm observed. Excitation of the sample at this wavelength leads to typical Yb³⁺ emission peaking at 980 nm. The only excited 4f level of Yb³⁺ is the ${}^2F_{5/2}$ level, so the broad band at 250 nm is caused

by charge transfer excitation. The peak of this charge transfer emission is blue shifted compared to Eu³⁺ reflecting the higher reduction potential of Yb³⁺ compared to Eu³⁺.

The emission and excitation spectra of LaPO₄ particles co-doped with 5% of each of Yb³⁺ and Er³⁺ are shown in Figure 18.

The absorption lines of Er³⁺ and a broad band peaking at 250 nm are visible in the excitation spectrum. The presence of this broad band is caused by charge transfer excitation of Yb³⁺ followed by energy transfer to Er³⁺. When the sample was excited in the broad band at 250 nm emission at 980 nm from Yb³⁺ and at 1530 nm from Er³⁺ was observed. The appearance of Yb³⁺ emission means that energy transfer is not complete. The direct excitation of Er³⁺ at 378 nm leads to Er³⁺ emission at 1530 nm as expected, but also Yb³⁺ emission at 980 nm is observed, showing that energy transfer from Er³⁺ to Yb³⁺ is also possible. This reverse process needs phonon energy from the surroundings because the ⁴I_{11/2} energy level of Er³⁺ lies a bit lower in energy than the ²F_{5/2} level of Yb³⁺. In LaF₃ particles energy transfer between Yb³⁺ and Er³⁺ is also observed after direct excitation of Yb³⁺ into the ²F_{5/2} level around 980 nm. In this matrix no charge transfer band is observed, similar to Eu³⁺. Particles of LaF₃ codoped with various amounts Yb³⁺ and 5% Er³⁺ were synthesized and the luminescence properties were determined in order to optimize the energy transfer.

The emission spectra of LaF₃ particles doped with a varying amount of Yb³⁺ and 5% Er³⁺ excited at 940 nm are shown in Figure 19. The emission spectra are normalized to the Yb³⁺ emission. Excitation at 940 nm was chosen because Er³⁺ has no absorption at this wavelength, so all emission from Er³⁺ is formed through sensitized

emission by Yb³⁺. At 1530 nm, Er³⁺ emission is clearly observed, proving the energy transfer between the ions. The Yb³⁺ emission at 980 nm is still dominating, so energy transfer is not complete. By increasing the amount of Yb³⁺ the relative Er³⁺ emission increases, so the energy transfer becomes more efficient up to a doping concentration of 20% Yb3+ where the maximum energy transfer is reached. Increasing the concentration of Yb3+ above 25% does not increase the energy transfer further. At increasing Yb³⁺ concentration the energy transfer becomes more efficient because the average distance between the ions decreases. It levels off when all the Er3+ ions have at least one Yb3+ ion as the nearest neighbor. The excitation spectra of Er3+ luminescence at 1530 nm for samples doped with different amounts of Yb3+ are also shown in Figure 19. The excitation spectra are scaled to the Er³⁺ absorption peak at 650 nm, because the amount of Er³⁺ in the sample is constant. It can be seen that the excitation peak of Er³⁺ in the sample without Yb³⁺ around 980 nm is very low and that it can be increased by co-doping with Yb³⁺. Increasing the amount of Yb³⁺ leads to a linear increase of the Yb³⁺ peak compared to the Er³⁺ peak, because more energy is absorbed by Yb³⁺ ions. It was reported in the literature that energy transfer from Yb³⁺ to Er3+ in Y2O2S nanoparticles is less efficient than in bulk material, because of the absence of low energy phonons. These low energy phonons are required to take up the small energy released when the excitation energy is transferred from Yb³⁺ to Er³⁺. In the present case, in which the nanoparticles are capped with organic molecules these low energy phonons are present in the form of vibrational and rotational energies, thus facilitating a more efficient energy transfer.

Example 11:

Decays of near-infrared emitting ions. Luminescence decays of NIR emitting ions could also be fitted using the model described above. Some decays with fits are shown in Figure 20A&B. Table 3 gives the values of the fit constants τ_R and C.

Table 3: Fit parameters for the NIR emitting ions in the two different particles.^a

Particle	Solvent	τ_{R} (μ s)	$C(s^{-1})$	τ _{av} (μs)
LaPO ₄ :Nd	CD ₃ OD	90	107	68
LaF ₃ :Nd	CH ₂ Cl ₂	347	23.5	265
LaPO ₄ :Er	CD_3OD	315	4.66	304
LaF ₃ :Er	CH ₂ Cl ₂	236	3.29	226
LaPO ₄ :Pr	CD_3OD	4.0	57.3	3.7
LaF ₃ :Ho	CH ₂ Cl ₂	35	27	31

a) All doping concentrations were 5%. The errors in τ_R are \approx 3% and in C \approx 10% determined

The luminescence lifetimes reported for Er^{3+} and Nd^{3+} are very long compared to values reported previously in organic environment. An increase in the luminescence lifetime by a factor 10-1000 is observed. For Nd^{3+} the values of the average lifetime and τ_R are higher for LaF_3 than for $LaPO_4$ reflecting the difference in the phonon energies of the matrices. This is the other way around for Er^{3+} . Both the value of τ_R and the average lifetimes are lower for the LaF_3 :Er particles compared to the $LaPO_4$:Er particles. Not only is the quenching stronger, but the lower τ_R indicates that there are more quenching possibilities for Er^{3+} remaining inside the particles. This

can be a result of the higher reaction temperature for the LaPO₄ particles, which functions as a better annealing step compared to the LaF₃ particles. The average lifetimes of Pr^{3+} and Ho^{3+} are in the μ s range reflecting the shorter radiative lifetimes of these ions. For Pr^{3+} in a phosphate host, a lifetime of 70 μ s for the 1D_2 level was reported in the literature and for Ho^{3+} in LaF₃, a lifetime of the 5F_5 level of 780 μ s was calculated.

Using the described model, different factors influencing the luminescence lifetimes of these particles were investigated. Er^{3+} usually shows a large dependence on the doping concentration and when the particles are doped with different concentrations of Er^{3+} this is indeed reflected in the values of τ_r and C (Table 4).

Table 4: Fit parameters for LaF_3 : Er and $LaPO_4$: Er at different doping concentrations.^a

	LaF ₃ :Er		LaPO ₄ :Er	
Concentration	$\tau_R(\mu s)$	C(s ⁻¹)	τ _R (μs)	C(s ⁻¹)
5%	236	3.29	315	4.66
2%	332	3.73	758	4.27
1%	465	3.54	1070	2.92

a) The errors in τ_R are \approx 3% and in C \approx 10% determined from duplicate measurements

A decrease in Er^{3+} concentration should not change the quenching constant C because the Er^{3+} ions are still in the same crystal site and at the same average distance to the surface. τ_R should increase with a decreasing doping concentration because

quenching processes like cross-relaxation and up-conversion become less important. This is clearly observed in the fitted values for LaF₃:Er with different Er³⁺ concentrations. A pronounced increase in τ_R is observed when the concentration is lowered, while the C stays about the same for the concentrations used here. For the LaPO₄:Er particles the constant C does change and becomes lower at a lower doping concentration. The different surface termination of the particles can play a role here. One quenching path is the transfer of the excitation energy of ions in the core of the particles to ions closer to the surface followed by quenching. This process is dependent on the concentration of the doping ion, because energy transfer becomes more efficient at higher doping concentrations, but also on the surface termination. In the LaF₃ particles lanthanide ions are coordinated directly to the ligand and other possible quenchers, but in the case of the LaPO₄ particles PO₄ groups are at the surface. In addition, quenching by solvents with a different quenching strength can be described using this model. Fit values obtained for LaPO₄:Nd and LaPO₄:Eu in methanol with a different degree of deuteration are given in Table 5.

Table 5: Fit parameters for LaPO₄:Nd and LaPO₄:Eu in methanol with different degrees of deuteration.^a

	Nd ³⁺		Eu ³⁺	
Solvent	τ _R (μs)	C(s ⁻¹)	$\tau_{R}(ms)$	C(s ⁻¹)
CH ₃ OH	87	733	7.0	4.06 10 ⁻²
CH ₃ OD	91	214	6.9	1.18 10 ⁻²
CD ₃ OD	90	107	6.9	1.31 10 ⁻²

The errors in τR are $\approx 3\%$ and in $C \approx 10\%$ determined from duplicate measurements

For all particles the same τ_R is found in agreement with the fact that the particle itself does not change. The C constants are different, as expected. A small quenching constant is found for the Eu³⁺ ions, because the energy gap of Eu³⁺ is relatively large and it is therefore less sensitive to quenching. In non-deuterated methanol the strongest quencher is the OH group and replacing this with an OD group leads to a large decrease in C. Replacement of the CH group by CD leads to another small decrease in quenching in the case of Nd³⁺, but in the case of Eu³⁺ the effect is negligible consistent with the larger energy gap between the excited state and the ground state for Eu³⁺.

One factor not included in the model is the difference in refractive index between the core and the surrounding solvent. The radiative lifetime of the lanthanide ion is dependent on the refractive index of the surroundings and it was previously found that particles in solvents with a different refractive index show a different luminescence lifetime. The average lifetime for LaF₃:Eu particles dissolved in different solvents increased from 6.1 ms in a refractive index of 1.525 to a average lifetime of 7.3 ms in a solvent with a refractive index of 1.352. The effect of the refractive index on the luminescence lifetimes of a bi-exponential fit was primarily seen on the long component of the two lifetimes, not on the short one. The short component is mostly determined by surface quenching and for this reason the effect of the refractive index on the results of the model will be small, except for changing the τ_R .

Another refinement of the model will be to take into account the size and size distribution of the particles. The size and size distribution is currently taken up by the constant C, averaging all the sizes in one batch. By taking into account the size and

size distribution we could look systematically at the effect of different particle sizes on the luminescent lifetime.

Example 12:

Surface modification of LaF₃ nanoparticles. One way of substituting a ligand for another ligand is by an exchange reaction. The LaF₃ nanoparticles were synthesized with ligands 1 and 2 in Figure 21 as described above. These ligands have a dithiophosphate headgroup, which binds mildly to the nanoparticle surface, to allow the growth of the nanoparticles and coordinates strong enough to prevent aggregation of the nanoparticles. Oxygen bearing coordinating groups have a higher binding affinity towards lanthanide ions and examples of these ligands would be dioctadecyl phosphate and oleic acid (ligands 3 and 4) in Figure 21, with a phosphate group or a carboxylic acid group.

The ligands used in the synthesis were exchanged by stirring a nanoparticle solution with ligand 3 or 4 and a small amount of base to ensure the deprotonation of the coordinating group. The exchange reaction of these ligands on the nanoparticles can be monitored using ¹H NMR. Figure 22 shows the result of nanoparticles synthesized with ligand 2 and exchanged with the phosphate ligand 3.

The ¹H NMR of the nanoparticles obtained after the synthesis show broadened signals of the ligand due to coordination of the ligand to the nanoparticle surface. This broadening of the NMR signals of nanoparticle bound ligands can be ascribed to the inhomogeneous distribution of the magnetic environment around the nanoparticle and a reduction in rotational freedom of the ligand. No free ligand is present, because no

sharp signals are observed for unbound ligand in solution. A complete exchange of the dithiophoshate ligand is accomplished as can be seen in the ¹H NMR spectrum of the exchanged product. The aromatic proton signals at 6.8 and 7.2 ppm of the dithiophosphate ligand have disappeared and the triplet of the CH3 terminal group of the phosphate ligand is present at 0.9 ppm. The ligand exchange is accompanied by a small change in solubility. Before the exchange the nanoparticles were not soluble in apolar aliphatic solvents like hexane or pentane, but after the exchange they were. This is a confirmation that before the exchange the surface of the nanoparticles consists of the aromatic groups of the ligand making the nanoparticles soluble in aromatic solvents like toluene. After the exchange the surface of the nanoparticles consists of the CH₃ groups making the nanoparticles also soluble in hexane and pentane. Figure 23 shows the ¹H NMR spectrum of LaF₃ nanoparticles synthesized with ligand 1, before and after the exchange with an excess of oleic acid. The ligands all contain long alkyl chains and in the proton spectrum these are mainly located between 0.9 and 2 ppm. It is difficult to observe changes in this region of the spectrum except for changes in the integrals. Protons next to the functional groups are shifted downfield and can be used to follow the exchange reaction.

The left spectrum shows the nanoparticles with ligand 1. This spectrum has a characteristic double triplet at 4.1 ppm of the protons next to the phosphor atom, but on the nanoparticles the signal is broadened and only a single broad peak is observed, as can be seen in the enlargement. The spectrum on the right shows the broadened signals of the oleic acid ligand coordinated to the surface of the nanoparticles after the exchange reaction. The exchange is complete, because the signal of ligand 1 at 4.1 ppm is absent and a peak of oleic acid at 5.2 ppm of the protons on the double bond is

observed. These results confirm that the ligand used in the synthesis of LaF₃ nanoparticles can be exchanged completely for another ligand with a stronger binding coordinating group. This provides a straightforward method for the introduction of functional groups at the surface of the nanoparticles. The stability of the nanoparticles was not much influenced. Apparently the coordination of the ligand to the nanoparticle surface is not the limiting factor for the stability, but the size of the hydrocarbon tails is.

Another possibility to functionalize the surface of the LaF₃ nanoparticles is to carry out the synthesis with ligands that already have functional groups. This is probably the most straightforward way to tune the solubility of the nanoparticles in different solvents. A synthesis route to obtain functionalized dithio ligands is shown Figure 24. The synthesis of these ligands is similar to the synthesis of ligand 2, using a functionalized phenol instead of phenol. Every functionality that does not interfere in the second step of the reaction can be used, making this a versatile method to functionalize the surface of the LaF3 nanoparticles. Nanoparticles synthesized with these ligands do show different solubilities. Nanoparticles synthesized with ligand 2 are soluble in apolar solvents like, toluene, dichloromethane, and chloroform. Nanoparticles synthesized with ligand 5 are soluble in solvents containing ester and ketone groups like, ethylacetate, cyclopentanone and γ -butyrolactone. The last two solvents are commonly used for the spin-coating of polymer solutions. The optical properties of the nanoparticles are independent on the functionalities of the ligand, so the functionalities do not interfere in the formation of the inorganic core of the nanoparticles. The same luminescence

spectrum and luminescence lifetime was measured and no change in the nanoparticle cores was observed with TEM.

Example 13:

Surface modification of LaPO₄ nanoparticles. The surface of LaPO₄ nanoparticles consists of phosphate groups and some bound ethylhexyl chains left after the synthesis of the nanoparticles. The surface of the nanoparticles is covered with OH groups of the phosphate groups, which provide the opportunity to covalently bind molecules to the surface. The presence of the OH groups was proven with solubility experiments. Directly after the synthesis, the nanoparticles were soluble in polar aprotic solvents like DMF and DMSO. This means that the surface has a polar nature, which is not consistent with a full coverage with ethylhexyl chains of the coordinating solvent. A full coverage with these apolar ligands would make the nanoparticles soluble in apolar solvents, like hexane, toluene, and dichloromethane. The nanoparticles can be solubilized in alcohols, such as methanol and iso-propanol by the addition of tetraalkyl ammonium hydroxides. The addition of tetraalkyl ammonium bromides does not lead to solubility in the alcohols, proving that the base is necessary to deprotonate the surface OH groups. Deprotonation of the surface OH groups leads to a negatively charged surface, that makes the coordination of the positively charged ammonium ions possible. For a functionalization reaction the OH surface groups can be reacted with alcohols, leading to phosphate esters. The OH groups of the surface phosphate groups were converted into chlorides by a reaction with phosphorus oxychloride. The surface of the nanoparticles can subsequently be functionalized by a reaction with an excess amount of dodecanol. The ¹H NMR of this product is shown in Figure 25.

The ¹H NMR signals of the dodecane chains are broadened due to the binding to the nanoparticles surface. The LaPO₄ nanoparticles before surface functionalization were only soluble in polar aprotic solvents like DMSO and DMF, but after the surface modification the nanoparticles are soluble in apolar solvents like toluene, chloroform and dichloromethane. The reaction of phosphorus oxychloride with the nanoparticles is a rigorous method to introduce the chlorides to the surface of the nanoparticles and characterization was performed to show that the nanoparticle core is still intact. TEM investigation of the reaction product showed that the nanoparticle cores are still intact and that no change in nanoparticle size and size distribution has occurred. Elemental analysis confirmed the presence of about 10 % organic material in the product, which is consistent with a monolayer formation around 5 nm nanoparticles. Luminescence studies also revealed no change in the environment of the Eu³⁺ ion when the surface modification is performed on Eu³⁺ doped nanoparticles as shown in Figure 26. The luminescence spectra have the same line positions and the same peak splitting showing that the Eu³⁺ ion is still in the LaPO₄ matrix.

This reaction can thus be used to make the nanoparticles hydrophobic, making them soluble in a-polar solvents. The reaction also offers a way to introduce other functional groups on the surface of the nanoparticles, making the tunability of the solubility properties possible. Haase *et al.* reported the surface modification of these nanoparticles by heating them in dodecylamine. Nanoparticles with dodecylamine coordinated to the surface were also soluble in apolar solvents and could be precipitated from solution by the addition of methanol.

Example 14:

Synthesis of core-shell nanoparticles of LaF₃. The synthesis of core-shell nanoparticles can be an easy way to enhance the luminescence properties of lanthanide doped nanoparticles, because interaction with the high-energy vibrations of the surrounding organic environment can be reduced in this way. A schematic representation of such a core-shell nanoparticle is shown in Figure 27. All the lanthanide ions involved in the luminescence of the nanoparticle are doped in the core and therefore have no direct interaction with the surface of the nanoparticle.

Above, we showed that the sizes of LaF₃ nanoparticles could be manipulated by changing the ratio of the ligand and reagents that form the core. This is a strong indication that the nanoparticles form after an initial nucleation followed by growth after the addition of more monomer. This mechanism allows the growth of a coreshell structure in one single synthesis step, by a stepwise addition of reagents. In the core-shell experiments, ammonium fluoride was used instead of sodium fluoride as the fluoride precursor, because this compound shows better solubility in water, making it possible to add fluoride precursor to the reaction mixture dissolved in a small amount of water. Core-shell nanoparticles with a core of La_{0.4}F₃Ce_{0.45}Tb_{0.15} and a shell of LaF3 were synthesized to study the influence of the shell on the quantum yield of the nanoparticles. During the growth of the shell no additional nucleation should occur, resulting in the formation of new un-doped nanoparticles. The size of the nanoparticles is a good indication for this. The size of the nanoparticles should increase roughly corresponding to an increase in volume that is similar to the amount of shell material added. The nanoparticle sizes before and after the growth of an undoped shell around a LaF₃CeTb core were measured using TEM. Figure 28 shows

typical TEM pictures of core nanoparticles and core-shell nanoparticles, with the corresponding size distribution histograms. A growth of the nanoparticles is clearly observed from 7.0 nm with a size distribution of 28.1% for the core nanoparticles, to 9.0 nm with a size distribution of 22.7% for the core-shell nanoparticles, when the same amount of material was used for the shell as for the initial core. This increase in size corresponds to a two times increase in volume, indicating that all of the added monomer has grown on the existing core nanoparticles and no new nucleation has occurred. This is consistent with the finding in above where nanoparticles with different sizes were synthesized by changing the amount of ligand. The size distribution is the same in absolute value but because of the increased size it looks like it has narrowed. In this case the effective ligand ratio is also lowered after the addition of the shell monomer, making it possible to form larger nanoparticles.

To determine the effect of the undoped core around the nanoparticles, the quantum yields of Ce^{3+} and Tb^{3+} doped nanoparticles were determined by comparing the emission intensity with the emission intensity of quinine bisulfate. Upon excitation of the Ce^{3+} absorption band at 282 nm, emission intensity of Ce^{3+} at 350 nm and four emission peaks of Tb^{3+} at 489, 543, 584, and 621 nm of the $^5D_4 \rightarrow ^7F_J$ (J=3-6) were observed. The quantum yield of the emission was determined by measuring both the emission of Ce^{3+} and Tb^{3+} . Three methods were used to grow a shell of LaF_3 around the core nanoparticles, which are described in the experimental section and differ in the way the shell reagents were added to the reaction mixture.

Table 6: Luminescence quantum yield of LaCeF3:Tb core shell nanoparticles.

Nanoparticles	Quantum Yield (%)
No shell	24.4
Method 1	42.7
Method 2	41.7
Method 3	54.2
Control ^a	27

a) In the control experiment the nanoparticles are synthesized in the same size as the core-shell nanoparticles, but with out the core-shell structure

The quantum yield of the core nanoparticles was very similar to the quantum yield determined above synthesized with NaF as the fluoride precursor, as an indication that the source of the fluoride ion has no influence on the optical properties of the nanoparticles. All methods to synthesize shell material were successful according to the quantum yield measurements. The best method was by growing the shell material by the alternate addition of small portions of the shell monomer. The quantum yield of nanoparticles coated in this way more than doubled. As a control experiment nanoparticles of the same size as the core-shell nanoparticles were synthesized, but with the core and shell mixed together to check if the increase in quantum yield is not caused by the dilution of the luminescent ions. Nanoparticles synthesized in this way had a quantum yield of 27%, only a little higher than the core nanoparticles, but clearly much lower than the core-shell

nanoparticles. The Ce³⁺ emission in these diluted nanoparticles was stronger compared to the core and the core-shell nanoparticles indicating that energy transfer from Ce³⁺ to Tb³⁺ is less efficient due to an increase in the average distance between the Ce³⁺ and Tb³⁺ ions. From this experiment it can be concluded that the increase in quantum yield is a result of the core-shell structure and not of nanoparticle size or a different distribution of the lanthanide ions in the nanoparticles.

Figure 29 shows the luminescence decay curves of both luminescent levels of the Eu³⁺ ion in LaF₃ nanoparticles, with different amounts of shell material. The non-exponential decay of the ⁵D₀ level is clearly observed and when one excess of shell monomer was grown on the nanoparticles a large change in the decay was observed. The decay becomes almost single exponential and an increase in the luminescence lifetime was observed. The values obtained for the luminescence lifetime after a bi-exponential fit are shown in Table 7. The increase in luminescence lifetime shows the increased shielding from quenching groups at the surface and the shift in the two components of the luminescence lifetime shows the decay was almost mono-exponential after the growth of a shell around the core.

Table 7: Values of the bi-exponential fit on the decays of the 5D_0 and 5D_1 level of LaF_3 : Eu nanoparticles with different amount of shell material in toluene.

	$^5\mathrm{D_0}$		$^{5}\mathrm{D_{1}}$		
	$ au_1$	τ_2	τι	τ_2	
no shell	7.9 (74%)	3.1 (26%)	2.0 (73%)	0.65 (27%)	
1 fold excess	9.7 (5.8%)	5.3 (42%)	2.3 (83%)	0.72 (17%)	
2 fold excess	9.9 (54%)	5.6 (46%)	2.4 (81%)	0.78 (19%)	
3 fold excess	9.5 (69%)	5.4 (31%)	2.5 (82%)	0.80 (18%)	

a) The percentages reflect the amount of the component contributing to the total lifetime.

The biggest change in the decay is observed after the addition of one excess of the shell material. The growth of more shell material has a smaller influence on the decay curve, but with the growth of a 3-fold excess a small rise in the decay curve is observed. This could be a result of reduced quenching of the higher Eu³⁺ levels like the 5D_1 , 5D_2 , or 5D_3 levels. The fitting of this decay was started after a 5 ms after the excitation pulse due to the rise in the decay. As a result the values of the luminescence lifetime are lower than for the 2-fold excess. The effect of the shell on the luminescence lifetime of the 5D_1 decay is also shown in Figure 29 and the corresponding values after a bi-exponential fit in Table 7. The effect is similar to what was observed for the 5D_0 level. The growth of one shell is enough to increase the luminescence lifetime and a thicker shell only has a small influence on the

b) The absolute errors in the fit are 0.2 ms in the lifetime and 5 % in the component contributing to the total lifetime.

luminescence lifetime. In the emission spectrum the emission from the ⁵D₂ level is located between 460 and 520 nm. In the core nanoparticles, no emission from this level could be observed, as shown in Figure 30.

After the growth of LaF₃ shell material emission lines start to appear at 464, 472, 489, and 509 nm of the ${}^5D_2 \rightarrow {}^7F_J$ (J = 0-3) transitions. These lines slowly become stronger after the growth of more shell material, indicating reduced quenching of the 5D_2 level. In the bulk material emission from the 5D_2 level is also observed at this concentration and temperature.

Conclusions

Several methods were developed to functionalize the surface of LaF₃ and LaPO₄ nanoparticles. This functionalization has an important influence on the solubility of the nanoparticles, which is important for the processibility of the system. The synthesis of core-shell nanoparticles for LaF₃ was successful, because the sizes of the nanoparticles were increased after growing a shell of undoped LaF₃ around the original core. The luminescence properties of the lanthanide ions were improved, which was demonstrated by the increase in quantum yield of LaF₃CeTb nanoparticles and the increase in luminescence lifetime of the emissive levels of the Eu³⁺ ion. The results show that it is possible to reduce the amount of quenchers coordinated to the lanthanide ions by the growth of an inorganic shell around the luminescent core.

Example 15:

Polymer waveguide amplifier doped with LaF3:Nd nanoparticles.

PMMA waveguides

PMMA waveguides doped with 10% w/w LaF₃:Nd nanoparticles were made by spin-coating a solution of PMMA and the nanoparticles in chlorobenzene on a silicon wafer, coated with a layer of 3.2 µm of SiO₂. Ridge structures with a height of 1.45 µm were etched in the 3.3 µm thick polymer layer using reactive ion etching (RIE). The formation of a gray film over the substrate indicates that redepostion of material, probably lanthanide oxides, takes place during RIE. The method of RIE relies on the reaction of highly reactive ions and radicals to form volatile products that can be pumped away from the vacuum chamber. No volatile products can be formed with the lanthanide ions in this process, leading to the redepostion of the lanthanide salts. An optical microscope image of waveguide structures made in this way is shown in Figure 31. It is clearly visible that the surface of the polymer next to the ridge structures is rough due to the redeposition, but the ridge structures itself seems to be smooth. The final device had air as the top cladding leading to a large difference in refractive index of the polymer (~1.49) and the top cladding.

The endfaces of the polymer waveguide can be made by cleaving the wafer or by dicing the wafer. Due to the weak adsorption of the polymer to the substrate, it was not possible to dice the wafer without destroying the polymer layer and cleaving of the substrate was the only possibility. Low loss endfaces can be made using the cleaving method, but the result is not very reproducible. The PMMA wafer contained waveguides with different width, from 10, 5, 4, 3, 2.5, 2 µm. The waveguides of 4

μm and less did not transmit any light at 1300 nm. The 5 μm channel was single-mode at the signal wavelength of 1319 nm and multi-mode at the pump wavelength of 578 nm, but the losses of this waveguide were substantial (18 dB at the signal wavelength). An excitation wavelength of 578 nm was chosen, because at this wavelength the Nd^{3+} ion has the strongest absorption peak. The measurements were performed on the 10 μm wide channels, which were multi-mode, both at the signal and pump wavelength. Single mode optical fibers with a core diameter of 9 μm were used to couple the light into and out of the waveguide. The total chip losses (propagation and coupling losses) at the pump wavelength were 12 dB on the 10 μm wide channel and at the signal wavelength the losses were about 11 dB. The calculated coupling losses for the signal beam were at least 6 dB, so on the 3 cm device the propagation losses were in the order of 2 dB/cm. These relatively high propagation losses are probably caused by the rough sidewalls of the device.

The continuous pump beam was butt-end coupled into the waveguide and the emission was collected at the end of the waveguide. The collected emission spectrum is shown in Figure 32. The emission spectrum of the waveguide is compared to the emission spectrum of LaF₃:Nd nanoparticles in dichloromethane solution. The emission peaks of the ${}^4F_{3/2} \rightarrow {}^4I_{11/2}$ and ${}^4F_{3/2} \rightarrow {}^4I_{9/2}$ transitions of Nd³⁺at 1060 and 1330 nm were observed in both spectra.

The emission peak at 880 nm of the ${}^4F_{3/2} \rightarrow {}^4I_{13/2}$ transition was not observed in the waveguide emission spectrum, because an isolator was used to filter out the pump light and this isolator also filtered out the 880 nm emission. The intensity ratio between the other two emission peaks was also different, because the isolator filtered

out part of the 1060 nm emission. A sharp decrease in the 1330 nm peak is observed above 1340 nm, because of the overtone absorption of CH bonds in the waveguide. Amplification measurements on the waveguides were done using a setup as shown in Figure 33. In order to measure amplification of the signal beam, the pump and signal light have to be combined before going into the waveguide. The pump and signal light were combined into a 9 μm core single mode fiber using a WDM and butt-end coupled into the waveguide. The intensity of the 1319 nm signal laser was unstable, so the pump light was chopped in order to measure the difference in signal intensity when the pump light is on and when the pump light is off. This configuration is only possible when all of the pump light is filtered out before the detector, or residual pump light will interfere with the amplified emission. The use of an isolator was found sufficient to filter out all of the pump light. The pump beam was chopped at 80 Hz for the amplification measurements and combined with the signal light coupled into the waveguide. The light coming out of the waveguide was collected using a multimode optical fiber with a core diameter of 50 μm and fed to a detector. The detector signal was shown on an oscilloscope. At a pumppower of 12 mW coupled into the device and a signal power of about 1 μ W, a signal amplification of 0.3 dB was measured.

The gain showed a dependence on the pump wavelength between 570 and 590 nm, with the maximum gain corresponding to the absorption peak of the Nd³⁺ ion at 578 nm. This amplification is equal to a signal increase of 10 %, using the equation below for calculation from mW to dB, with I_{in} the signal intensity without pump light and I_{out} the signal intensity with pumplight, both in mW.

$$dB = 10 \times log \left(\frac{I_{out}}{I_{in}}\right)$$

This signal increase measured with the detector could also be a result of spontaneous emission of the Nd³⁺ ions. This can be studied by turning off the signal laser and measuring the intensity differences using a chopped pump laser. The effect of the spontaneous emission on the detector signal was approximately 1 % of the amplification signal, so the effect of spontaneous emission can safely be ignored.

Considering the large coupling and propagation losses of the device net gain will not be possible in this waveguide. To decrease the coupling and propagation losses other waveguides were made using another polymer. The coupling losses can be decreased using a device structure that better fits the fiber light mode with the polymer waveguide mode and the propagation losses can be lowered using other techniques to make the waveguide structures, instead of RIE.

Example 16:

SU-8 waveguides

Low loss waveguides of SU-8 have been fabricated. Nanoparticle doped SU-8 waveguides were made by mixing a LaF₃:Nd nanoparticle solution in γ -butyrolactone with a commercially available SU-8 solution in γ -butyrolactone. SU-8 is a negative photoresist that can be patterned by photolithography, without the use of RIE. The structures of the molecules in this photoresist solution are shown in Figure 34.

The monomer consists of a bisphenol A epoxy resin that can be polymerized by acid catalysis. A sulfonium ion is used as the photoinitiator for the polymerization, which

generates protons upon UV irradiation. The solvent used for the photoresist composition is γ -butyrolactone, of which the structure is also shown in Figure 34. Due to the high refractive index of the polymer of 1.59, glass can be used as the substrate. The polymer shows good adhesion to glass, making it possible to dice the endfaces. Due to the strong cross-linking of the polymer it is also possible to apply a cladding over the device, without destroying the waveguide structures.

SU-8 layers doped with 20 % w/w of the LaF₃:Nd nanoparticles were spincoated on 10 cm borosilicate wafers. The processing of the doped SU-8 was similar to the processing of the un-doped photoresist and only a few modifications were necessary. First of all, the temperatures at which the spin-coated layers were handled before the development had to be lowered compared to the SU-8 without nanoparticle doping. Temperatures above 75 °C lead to desorption of the ligand from the nanoparticle surface, which has a negative effect on the solubility of the nanoparticles in organic solvent. When the doped layers were handled at temperatures exceeding 75 °C, a thin layer of insoluble nanoparticles remained on the substrate after development. For this reason the layers were heated at 65 °C for two hours after spin-coating to remove the solvent. After the development the layers were hard-baked at the usual temperature of 150 °C, because in the highly cross-linked polymer the nanoparticles will not aggregate anymore. Another difference for the doped SU-8 layers compared to the SU-8 layers without nanoparticle doping was in the irradiation time. For the layers without nanoparticle doping the irradiation time was 10 seconds, but for the doped layers an irradiation time of 300 seconds was necessary. There can be several reasons for this increase in irradiation time. Most of the solvent is removed after the soft-bake step, even at 65 °C, so it is unlikely that residual solvent is the reason for the increase

in radiation time. A valid reason can be the slight desorption of ligand from the nanoparticle surface in the softbake step. This ligand can act as a base during the crosslinking of the polymer, slowing down the reaction rate. Another possibility is that the photoinitiator is not randomly distributed in the polymer film. The ligand monolayer around the nanoparticles creates a hydrophobic film, which can cause an inhomogeneous distribution of the photoinitiator.

The waveguides were finished by spin-coating a layer of HEMA-styrene on top of the device as a cladding, followed by dicing the endfaces.

Optical microscope pictures of the waveguides with the cladding on top are shown in Figure 35. The left picture shows a top view of some channels covered with the cladding. The picture shows straight and smooth channels. The right picture shows a cross-sectional view of a 5 by 10 µm channel. It can be seen that the top of the channel is a bit wider than the bottom of the channel, which can be caused by the unoptimized irradiation conditions. During the irradiation the top of the polymer slab can heat up by the long wavelength components in the UV light source. Due to the higher temperature in the top layer the reaction rate was increased causing the top of the channel to be wider. After dicing the endfaces, the total length of the waveguide was 5.2 cm.

Optical pumping at 578 nm was not successful for the SU-8 device, because the polymer was not stable under high pump fluxes of this wavelength. Yellowing of the polymer at the high temperatures of the hard bake step is probably the cause of this

instability. The yellowing of the polymer is probably caused by the photoinitiator, leading to absorptions well into the visible part of the spectrum, even at 578 nm.

Another possibility to excite the Nd³⁺ ions is at 795 nm. The absorption of this peak is a bit weaker, but an advantage of this wavelength is that pump and signal wavelength are closer together, making it more easy to match the pump and signal light mode. Nd3+-doped fiber amplifiers are usually pumped at this wavelength for this reason and because a large variety of cheap pump-lasers is available at this wavelength. It is unlikely that the yellowing in the SU-8 polymer is also present at this wavelength. The optical fibers had to be connected to the device for the pumping at 795 nm, because the pump laser and a good alignment setup were not available at the same place. Optical fibers with a 9 µm core were connected to the device by using glue curable by UV light. A laser operating at 850 nm was used for the alignment and with the glue between the fibers and the device a minimum loss of 7 dB over the entire device for a 10 by 10 μm channel was measured, including the coupling losses. The propagation losses of this waveguide were in the order of 1 dB/cm or less, assuming coupling losses of 1-2 dB. These losses are very similar to a waveguide without nanoparticle doping, showing that the nanoparticles are well dispersed in the polymer and do not contribute significantly to scattering losses. The losses were increased to 12 dB at 850 nm after curing of the glue by UV irradiation. This increase in the losses can be a result of stress in the glue resulting from an inhomogeneous curing, or as a result of the instability of SU-8 towards UV light.

The chip losses of the waveguide were determined using a white light source and the loss spectrum is shown in Figure 36. The minimum loss of 8.5 dB over the entire

device is around 1100 nm and at the signal wavelength of 1319 nm, the loss is 11.5 dB. At 738, 798, and 863 the Nd³⁺ absorption lines are observed originating from the ground state level to the ${}^4F_{7/2}$, ${}^4F_{5/2}$, and the ${}^4F_{3/2}$ levels respectively. Below 700 nm the losses of the device increase strongly due to the yellowing effect of the polymer.

The minimum loss window around 1300 nm can clearly be observed, with the overtone absorptions of the CH bonds around 1200 nm and above 1400 nm. The high losses around 1500 nm show that this polymer is unsuitable for waveguiding at this wavelength. The emission spectrum of Nd³⁺ after pumping the waveguide at 795 nm with 50 mW of input power and collecting the emission at the end of the waveguide structure is shown in Figure 37. The different ratio of the 1060 and the 1330 nm emission peaks compared to the emission spectrum in solution was caused by the filters, the same as for the PMMA device. A small peak of the pump laser at 795 nm is still observed, but this small signal did not interfere with the amplification measurement. Amplification measurements were performed using the same setup as used for the PMMA device. The pump beam and signal beam were coupled in one 9 μm fiber using a 2x2 splitter for the amplification measurement. Due to the high losses originating from the splitter, only 5 mW of pumppower could be coupled into the device. Using a chopped pump beam and leading the detector signal into the oscilloscope, a signal amplification of 0.1 dB could be measured on the 10 by 10 μm channel of 5.2 cm length.

Example 17:

Modelling of the amplification

Several models have been described to calculate the optical gain in Nd³+-doped glass fibers. The simplified model described in the appendix can be used to calculate the optical gain that can be reached in the polymer waveguides. This model does not take into account the losses of the chip, the effect of excited state absorption (ESA), and amplified spontaneous emission (ASE). When these effects take place the measured optical gain will be lower than the calculated gain. The last two processes will deplete the excited state of the Nd³+ ion by the absorption of a signal photon (ESA), or by amplification of light at another wavelength than the desired signal wavelength (ASE). When these processes take place the actual population of the excited state will be lower than the calculated one. The Nd³+ scheme for optical amplification is a 4-level system because the ${}^4I_{13/2}$ level is not the ground state and non-radiative decay from the ${}^4I_{13/2}$ level to the ${}^4I_{9/2}$ ground state is very fast. As a result the ${}^4I_{13/2}$ is always empty and a population inversion is always present. Amplification now only depends on the amount of ions that are in the ${}^4F_{3/2}$ excited state.

Factors that influence the amplification are the concentration of the lanthanide ion and the length of the device. Not taking into account any losses in the device, these factors counteract each other. When the concentration of the lanthanide ion is doubled, the length of the device can be halved to have the same amplification. This is of course different when losses and the effect of concentration quenching are taken into account. In glass fibers the losses are very low and concentration quenching lowers the luminescent lifetime of the ions even at concentrations below 1%. As a

result, fiber amplifiers are very long, extending to several meters. For integrated devices the length of the amplifier has to be reduced in order to fit the amplifier on a small chip. As a consequence, the concentration of the lanthanide ion has to be increased in order to have sufficient amplification. The optical losses are usually a bit higher for polymers compared to glass materials, making the requirement for a shorter device even more important. Two other factors of the waveguide design that have an influence on the amplification are the pump power and the cross-section of the channel. The pump density (power/area) determines the amount of ions that are in the excited state. At a higher pump density more ions will be in the excited state. The luminescence lifetime is another important factor in the amplification scheme. It is a measure for the amount of time an ion spends in the excited state and at a fixed pumppower it determines directly how much of the population is in the excited state. The emission cross-section determines how much stimulated emission there will be. The higher the emission cross-section the more stimulated emission and the more amplification of the signal. The emission cross-section and the luminescence lifetime are related to each other and this ratio will be the best when the quantum yield is unity.

The calculated amplification for the PMMA and SU-8 waveguide channels are shown in Figure 38, for the pumppowers of 12 mW and 5 mW, respectively, and for the channel sizes used in the experiments. The PMMA device was only 3 cm long and at this length the maximum amplification is not yet reached. The calculated value of 0.25 dB for a 3 cm chip is lower than the measured value of 0.3 dB but this is probably within experimental error, considering the approximations in the calculation.

SU-8 waveguide had a length of 5.2 cm and at this length the maximum gain is almost reached. In this case the measured gain of 0.1 dB is also slightly higher than the calculated gain of 0.08 dB. The calculated values correspond very well with the measured values and further evaluation of the amplification results can be done theoretically. Higher gain can be obtained by improving the waveguide design and by coupling more pump light into the device. The theoretical optical gain of an SU-8 waveguide channel, with dimensions of 10 by 10 μm, as a function of the pump-power is shown in Figure 39. It is clearly observed that the gain increases substantially when the pump-power is increased from 5 mW to 100 mW. A pump-power of 100 mW is often reported for amplification measurements and should be obtainable under optimized conditions.

The length of the device, or the concentration of Nd³⁺ ions in the waveguide have to be increased in order to measure the maximum obtainable gain at these higher pump powers. Another large performance increase can be reached when the size of the channel is decreased as shown in Figure 40. Decreasing the size of the channel increases the density of the pump power, so the effect is similar to an increase in pump power. The pump power in these calculations was at a constant 100 mW and the effect on the optical gain is similar to increasing the pump-power. With channel dimensions of 4x4 µm the optical gain is close to 10 dB, which is similar to the gain obtained for a Nd³⁺-doped fiber amplifier. To observe this gain experimentally the length of the device has to be increased to about 25 cm, or the concentration of Nd³⁺ has to be increased 5 times.

In order to show that the measured gain is a result of the increase in luminescence lifetime of the Nd^{3+} ion in the nanoparticles, compared to organic complexes, the optical gain as a function of the luminescence lifetime is shown in Figure 41, for a 10 by 10 μ m waveguide channel and a pump power of 100 mW.

The amplification is limited to only 0.009 dB for a luminescence lifetime of 1 µs, which is generally observed for complexes with CH bonds close to the lanthanide ion. The highest luminescence lifetime of Nd³⁺ observed in organic systems so far was 9 μs and for this luminescence lifetime the optical gain is limited to 0.08 dB. For a luminescence lifetime of 200 µs as observed in the LaF₃:Nd nanoparticles the optical is 1.7 dB for this device. In these calculations it is assumed that the absorption and emission cross-sections of Nd³⁺ is the same in the different materials. Small differences in these values may be observed due to changes in symmetry, but these differences will not compensate for the large difference in optical amplification that is observed. It is very clear that with the luminescence lifetimes observed in organic complexes it will be difficult to measure the small gain. The increase in luminescence lifetime we obtained, by doping the Nd³⁺ ion in the LaF₃ nanoparticles is responsible for the fact that we were able to measure optical gain. The doping of Nd3+ ions into the LaF₃ nanoparticles leads to optical properties that allow amplification close to what is observed in Nd³⁺-doped glass fiber amplifiers.

Amplification measurements using other ions should also be possible. The Er³⁺ ion, which is commonly used in optical fiber amplifiers, also shows emission in the LaF₃ nanoparticles, but for amplification at 1530 nm other polymers have to be used to reduce the optical loss at the signal wavelength. The organic ligand that is protecting

the nanoparticles from aggregation also contains a lot of CH bonds, but the total amount of ligand is low in the whole system and a proof of principle should be possible for amplification by Er³⁺ ions as well with this ligand and a halogenated polymer.

Conclusions

Polymer waveguides were successfully doped with LaF₃:Nd nanoparticles. The emission spectra of Nd³⁺ show that the nanoparticles are in the waveguides and the optical losses of the waveguide indicate that the nanoparticles are not clustered and therefore have a small contribution to scattering in the waveguide. Optical gain was demonstrated. For the PMMA waveguides an amplification of 0.3 dB at 1319 nm at a pump power of 12 mW was measured and for the SU-8 waveguides an amplification of 0.1 dB at a pump power of only 5 mW was measured. Theoretical calculations show that increasing the confinement of the pump and signal beam and an increase in pump power should lead to an amplification of 8 dB for a 4 by 4 µm channel, which is very close to the amplification that is obtained in glass fiber amplifiers doped with Nd³⁺. It is shown theoretically that this amplification is due to the increase in luminescence lifetime compared to organic complexes and that amplification with the luminescence lifetimes of organic complexes is very limited. The use of these organically soluble nanoparticles is therefore a major advantage over the use of lanthanide complexes as the gain material in polymer hosts.